

# **For Reference**

---


**NOT TO BE TAKEN FROM THIS ROOM**



Ex LIBRIS  
UNIVERSITATIS  
ALBERTAENSIS







Digitized by the Internet Archive  
in 2022 with funding from  
University of Alberta Library

<https://archive.org/details/Gold1978>









THE UNIVERSITY OF ALBERTA

QUANTITATIVE METHODS IN THE EVALUATION OF THE QUATERNARY  
GEOLOGY OF THE SAND RIVER (73L) MAP SHEET, ALBERTA, CANADA.

by



CHRISTOPHER M. GOLD

A THESIS

SUBMITTED TO THE FACULTY OF GRADUATE STUDIES AND RESEARCH  
IN PARTIAL FULFILMENT OF THE REQUIREMENTS FOR THE DEGREE  
OF DOCTOR OF PHILOSOPHY

DEPARTMENT OF GEOLOGY

EDMONTON, ALBERTA

SPRING, 1978





## ABSTRACT

The Sand river map sheet, located in east-central Alberta, possesses several bedrock channels buried beneath Quaternary glacial deposits and known only from drill hole information. A total of 995 samples from 78 drill holes were analyzed for 22 compositional properties in an attempt to delineate the stratigraphy of the till units present. The measurements included granulometry, grain lithology, calcium carbonate equivalent and bulk (major and minor element) chemistry. A reliable, rapid, constant -volume carbonate analysis apparatus was developed for the project. Bulk chemistry was performed by fusing till samples into glasses and analyzing them using energy-dispersive electron microprobe techniques. The correction procedures required for quantitative energy-dispersive analysis are described in some detail, along with EDATA, a computer program for applying them. Corrections are applied for atomic number, absorption and fluorescence effects, instrumental calibration corrections, background (X-ray continuum) shaping and fitting as well as peak overlaps.

Problems are often encountered in contour mapping of irregularly spaced data points by computer, and new techniques have been devised and implemented in the TRIMAP package, based on subdividing the map area into triangles. Features include the use of triangular element data structures, a local homogeneous coordinate system, the





automatic generation of a triangular network, the selection of a suitable triangle optimization criterion and the choice of a suitable smooth interpolant over individual triangular domains.

Four distinct till units were distinguished in the study area on the basis of their compositions. A high carbonate middle till separates an upper and a two member lower till. The compositional parameters of these tills were categorized as textural, erratic or local. In the fine fraction of a till unit, vertical compositional variation was low by comparison with lateral variation, which appeared to be related to sub-ice topography. The glacial history proposed relates the four suggested ice advances to the varying regional drainage as indicated by the major bedrock channels. The regional topography at each stage was estimated. Compositional analysis of tills, as well as the topographic reconstruction of stratigraphic boundaries, is important in studies of glacial history.





## ACKNOWLEDGEMENTS

In a study of this nature, covering a wide variety of topics, it is inevitable that many people have provided advice and assistance. I am greatly indebted to Dr. J.A. Westgate , my original supervisor, for initially suggesting the topic of study, and, upon his departure, to Dr. R.W. May who inherited the unenviable task of continuing what he had not initiated, and without whose advice and encouragement the project would never have been completed. Drs. U.M. Maydell, F.W. Schwartz and G.D. Williams also provided much needed advice on occasion, and their help is greatly appreciated.

Much gratitude is also due to various members of Alberta Environment. Funding for routine laboratory analysis of the drill hole samples was provided by the Research Secretariat of Alberta Environment, and all the samples could not have been analyzed without their assistance and the dedicated help of Mrs. D. Guidera. Mr. H.A. Kerr of the Earth Science and Licensing Division has also been more than generous with his advice, time and patience in promoting and overseeing the laboratory project, as well as making the necessary drill hole samples available. Mr. T. Yoon, Mr. H. Vander Pluym and the field engineers of the Earth Science and Licensing Division have also contributed their time and labour, and I am duly grateful.

In addition to those mentioned above, acknowledgement



must be made of those whose contributions are an integral part of this project. In the work on contour mapping Dr. J. Ramsden of Alberta Research contributed many basic ideas on surface representation, and Mr. T. Charters of Alberta Environment developed some of the key concepts of triangular mesh generation. In the development of the electron microprobe correction procedures, the project was essentially the team work of Dr. D.G.W. Smith and Mr. D.A. Tomlinson of the Department of Geology, University of Alberta, and myself. While it is not possible to specify who thought of what technique in the many discussions spreading over several years, certain broad areas of responsibility can be identified. Dr. Smith's role was primarily to conceive of an analytical approach to the problem of energy dispersive microprobe analysis, identify problems and suggest general approaches to their solution. This is especially true with respect to the physics of X-rays. Mr. Tomlinson was primarily concerned with the implementation of the instrumentation needed for the analytic approach used. My own role has principally been the mathematical and computer - based implementation of the chosen method, as well as contributing to the identification of and definition of problem areas. However, many alternate strategies were attempted in the development of the current procedure, and all team members have been heavily involved in all aspects of their suggestion, evaluation and interpretation.

Finally, I would like to thank Mr. S. Launspach for his





assistance with routine microprobe analysis, Dr. Smith for access to the electron microprobe, and my own family for both their assistance in the preparation of the manuscript and their tolerance of the project.





## TABLE OF CONTENTS

CHAPTER	PAGE
ABSTRACT	iv
ACKNOWLEDGEMENTS	vi
1.0 INTRODUCTION	1
2.0 STUDY AREA	4
2.1 Location and Data Availability	4
2.2 Topography and Drainage	6
2.3 Previous Work	6
3.0 GENERAL LABORATORY METHODS	9
3.1 Types of Analyses	9
3.2 Sample Examination	9
3.3 Statistical Evaluation of Laboratory Methods	12
3.4 Grain Size Analysis	18
3.5 Lithological Analysis	19
3.6 Carbonate Analysis	20
3.7 Bulk Chemistry	22
4.0 CARBONATE APPARATUS	23
4.1 Analytic Methods	23
4.2 Equipment	24
4.3 Temperature Correction	26
4.4 Calibration	28
5.0 ENERGY DISPERSIVE MICROPROBE ANALYSIS	30
5.1 Principles of Electron Microprobe Analysis	30
5.2 Quantitative Electron Microprobe Analysis	32
5.3 Energy Dispersive Correction Procedures	33
5.4 Instrumental Calibration Corrections	34
5.5 Background Shape	36
5.6 Background Fitting	48
5.7 Peak Overlap	50
5.8 Calculation of Composition	55
5.9 Sequence and Implementation of Correction Procedures	57
6.0 AUTOMATED CONTOUR MAPPING TECHNIQUES	66
6.1 Conventional Contour Mapping Techniques	66
6.2 Triangular Element Data Structures and Coordinates	68
6.3 Data Structure Generation and Optimization	72
6.4 Choice of an Optimization Criterion	76
6.5 Mapping with Triangular Elements	79
6.6 Requirements of an Interpolating Function	81



# TABLE OF CONTENTS continued

6.7 Shape Functions	82
6.8 Contour Mapping with the Interpolant	85
6.9 Extension into Higher Dimensions	94
7.0 LABORATORY RESULTS	97
7.1 Differentiation of Till Units	97
7.2 Relation of Analyses to Till Units	100
7.3 Relationships between Analyzed Variables	103
7.4 Choice of Variables for Till Differentiation	112
7.5 Compositional variation within till units	113
7.6 Vertical Variation in Till Composition	114
7.7 Local depositional processes	116
7.8 Lateral Variation in Till Composition	117
7.9 Sampling and Statistical Model	117
7.10 Local Compositional Variation	130
8.0 PREGLACIAL GEOLOGICAL HISTORY	134
8.1 Cretaceous Stratigraphy	134
8.2 Tertiary Stratigraphy	134
8.3 Bedrock Channels and Preglacial Drainage	136
9.0 GLACIAL HISTORY	139
9.1 Lower Member of the Lower Till and Kikino Channel Formation	139
9.2 Deposition of the Upper Member of the Lower Till	144
9.3 Bedrock Thrusts	149
9.4 Middle Till Deposition	154
9.5 Upper Till Deposition	162
9.6 Present Topography	162
9.7 Historical Summary	166
9.8 Correlation of Surficial Units	168
10.0 CONCLUSIONS	172
10.1 Requirements for Stratigraphic Correlation	172
10.2 Choice of Compositional Measurements	172
10.3 Sampling Methodology	173
10.4 Regional Stratigraphy	174
10.5 Hydrology	174
10.6 Analytic and Display Techniques	175
REFERENCES	177
APPENDIX 1 - Formal Laboratory Procedures	186
APPENDIX 2 Calcium carbonate equivalence and grain size analyses for all analyzed samples.	191
APPENDIX 3 Grain lithology and bulk chemical analyses for all analyzed samples.	235
APPENDIX 4 Plots of each drill hole, showing all analyzed variables, lithology and stratigraphic interpretation.	279





## TABLE OF CONTENTS continued

APPENDIX 5 Analysis of covariance for each analyzed variable, comparing between - units and within - units variance.	358
APPENDIX 6 Principal components factor analysis of the 22 analyzed variables.	370
APPENDIX 7 Contour maps showing the lateral variation of sand, clay, grain lithology, calcium carbonate equivalent and bulk chemistry in each of the four till units.	374
APPENDIX 8 Between-units versus within-units analysis of variance for each analyzed variable.	439
APPENDIX 9 Analysis of variance for each analyzed variable, comparing variation between areas, till units and vertical positions within till units.	451





## LIST OF TABLES

Table	Description	Page
1	Legal locations of drill holes used in this study.	10
2	Results of analyses of test samples used to test laboratory procedures for bulk chemistry, grain size, grain lithology and calcium carbonate equivalence.	13
3	Analyses of variance of the bulk chemistry values of Table 2.	15
4	Analyses of variance of the grain size, grain lithology and calcium carbonate equivalence values of Table 2.	17
5	Overlap coefficients expressed as percentages of measured intensities in analysis peaks of overlapping elements.	53
6	Typical analyses obtained using the EDATA program.	63
7	Comparison of analyses performed by the EDATA program with the best available values.	65
8	Variation of analyses between till units.	102
9	Analyzed variables that differentiate between till units.	104
10	Significance levels of the factors and interactions of Appendix 9.	120
11	Tentative correlation of surficial units between Saskatchewan and Alberta.	170



## LIST OF FIGURES

Figure	Page
1 Index map of the study area.	5
2 Drill hole index and generalized topographic map of the study area.	7
3 Construction of carbonate analysis equipment.	25
4 Plots of pressure (x-axis) versus time (y-axis) for several typical materials, using the apparatus of Fig. 3.	29
5 Plots of $n$ versus continuum energy for various operating voltages, obtained from the ratios of copper and diamond spectra.	42
6 Logarithmic plot showing the dependence of $g$ on $E_0$ .	43
7 Plot showing the relationship of $c$ and $E_0$ .	44
8 Plot showing the relationship of $f(E_0)$ to $E_0$ .	45
9 Plots of the ratios of the Z-corrected spectra of assorted metals and oxides with Z-corrected carbon spectra.	47
10 Spectra of various compounds before (upper) and after (lower) continuum and escape peak removal.	51
11 Diagrammatic representation of the overlap coefficients from F to Zn of Table 5.	54
12 Sample EDATA printout for a standard.	56
13 Sample EDATA printout for a sample of unknown composition.	58
14 Schematic flowchart for the EDATA program.	60
15 Triangular element data structures.	69
16 Derivation and use of a local homogeneous coordinate system for a triangle.	70
17 Triangular data structure generation and optimization.	73
18 Rate of convergence of optimization of a triangular mesh.	75
19 Optimization criteria.	77
20 An example of triangulated contour mapping - a magnetic field map.	86
21 Plot of $Z = \sin(X) \sin(Y)$	87
22 Bedrock surface elevation in the map area, contoured using gridding and distance - weighted elevations at data points.	88
23 Bedrock surface elevation in the map area, using gridding and distance weighted best - fit planes at data points.	90
24 Bedrock surface elevation in the map area, using automatic triangulation.	91
25 Hand drawn bedrock surface elevation map.	92





# LIST OF FIGURES continued

26	Hand generated triangulation of the map area.	93
27	Bedrock surface elevation in the map area, using hand generated triangulation.	95
28	Furthest neighbour dendrogram showing the relationships between the 22 analyzed variables.	106
29	Unweighted pair group dendrogram showing the relationships between the 22 analyzed variables.	107
30	Dendrite analysis dendrogram showing the relationships between the 22 analyzed variables.	108
31	Mode analysis dendrogram showing the relationships between the 22 analyzed variables.	109
32	Plot of factor 2 in till unit 1.	122
33	Plot of factor 2 in till unit 2.	123
34	Plot of factor 2 in till unit 3.	124
35	Plot of factor 2 in till unit 4.	125
36	Plot of erratic / local cation ratio in till unit 1.	126
37	Plot of erratic / local cation ratio in till unit 2.	127
38	Plot of erratic / local cation ratio in till unit 3.	128
39	Plot of erratic / local cation ratio in till unit 4.	129
40	Bedrock geology of the Edmonton - Lac La Biche - Cold Lake area.	135
41	Thickness in feet of unconsolidated deposits immediately underlying till unit 4.	140
42	Elevation in feet of the base of till unit 4.	141
43	Thickness in feet of till unit 4.	142
44	Elevation in feet of the top of till unit 4.	145
45	Thickness in feet of unconsolidated deposits immediately underlying till unit 3.	146
46	Elevation in feet of the base of till unit 3.	147
47	Thickness in feet of till unit 3.	148
48	North - south cross - section through the Vilna bedrock thrust.	151
49	Fence diagram along the bedrock channels of the study area.	153
50	Elevation in feet of the top of till unit 3.	155
51	Thickness in feet of unconsolidated deposits immediately underlying till unit 2.	156
52	Elevation in feet of the base of till unit 2.	157
53	Thickness in feet of till unit 2.	158
54	Elevation in feet of the top of till unit 2.	159
55	Thickness in feet of unconsolidated deposits immediately underlying till unit 1.	160
56	Elevation in feet of the base of till unit 1.	161
57	Thickness in feet of till unit 1.	163
58	Elevation in feet of the top of till unit 1.	164
59	Present surface elevation in feet.	165



## 1.0 INTRODUCTION

For many years the glacial geology of the Prairies of Western Canada has been known primarily in two dimensions. This was because of the relatively infrequent occurrence of suitable vertical sections. Consequently, although the glacial stratigraphy over much of the Prairies might be expected to be relatively consistent and extensive, little regional information has been available below the uppermost units.

A general exception to this state of affairs has been the work of Christiansen (1968a,b, 1972a,b) utilizing information and samples from holes drilled in the process of regional hydrogeological surveys. He was able to define the regional glacial stratigraphy over much of central Saskatchewan by defining the stratigraphic relationship between drill holes on the basis of electric logs and sample examination. Sauer (1974) mentioned the practical value of till stratigraphy in highway engineering and water supply problems. In particular, where the tills differed in their engineering properties, embankment failure could have been caused by water seepage along inter-till boundaries, and major surficial aquifers were usually located between or below major till layers.

Visual examination of available vertical sections is of great value in conventional field work, but laboratory analyses are often very helpful in the identification of till units. When field examination is restricted to samples



taken from a hole during drilling operations, laboratory analysis is necessary to determine criteria for distinguishing superimposed till units.

The primary objective of this study was to examine the regional glacial stratigraphy in an area of Alberta where drill hole samples and data were readily available. The area was also of interest as three bedrock channels were known to be present, and it was hoped that some relationship could be observed between regional glaciation and drainage. It was therefore necessary to determine those compositional properties of greatest value in till identification as well as to determine efficient sampling and analytic strategies for any future work. In the pursuit of these objectives it became necessary to develop new analytic tools. The development of improved apparatus for measuring till carbonate content was one result (see Chapter 4), as was the development, in association with Dr. D. G. W. Smith, of fully quantitative energy dispersive microprobe correction procedures as used in electron microprobe analysis of the bulk chemistry of the fine fraction of till samples (Chapter 5). Finally, in the display of topographic data, grid-type computer contouring algorithms were found to be inadequate because of their inability to resolve features with few sample points, and the difficulty in obtaining a surface that passes precisely through all the data points. A new approach to contour mapping was consequently developed that honoured all data points instead of smoothing the surface,





as well as permitting the user-definition of features such as valley-lines.

Due to the variety of the subject matter covered, some explanation of the order of presentation may be applicable. After a brief description of the study area, the laboratory methods used in the examination of the drill hole samples are discussed, along with the statistical tests used to evaluate the various analyses. Because the carbonate and microprobe analysis techniques were developed especially for this study, the next two chapters discuss them in detail. In a similar manner, the display and interpretation of the greater than 20,000 analytic results required the availability of suitable display methods, and hence the chapter on contour mapping precedes the detailed examination of laboratory results. Finally, one of the primary purposes of the laboratory analyses is the delineation of the till units found. Thus an interpretation of the glacial history of the area requires in advance both a knowledge of the till units present, as well as some background knowledge of the preglacial history. Historical geology thus represents the synthesis of much previous work.



## 2.0 STUDY AREA

### 2.1 Location and Data Availability

In 1974 the Groundwater Branch of Alberta Environment published a report entitled "Buried Channels in the Edmonton - Lac La Biche - Cold Lake Area, Alberta," (Yoon and Vander Pluym, 1974). This was a geological study of three well defined buried bedrock channels: the Beverly, the Kikino and the Helina. Field work in the area included test drilling, lithologic and electric logging, and side-wall sampling of the drilled wells. One hundred and fifty two test holes were drilled, usually to bedrock, with sidewall samples taken from ninety eight of them. Additional well log information was obtained from several other sources, including oil and gas exploration wells.

In this study data were utilized from seventy eight of these drill holes in an area of good sample density and potentially interesting geology where three buried channels were present. All samples from these drill holes were examined and 995 were analyzed in the laboratory. Fig. 1 shows the location of the study area and its relation to the previous report. All drilled holes were located between townships 56 and 68, ranges 2 to 17 west of the fourth meridian. The study area falls mostly within the Sand River (73L) 1:250,000 map sheet of the National Topographic Survey, although a few drill holes are located just to the south (map 73E) or west (83I) of this area.





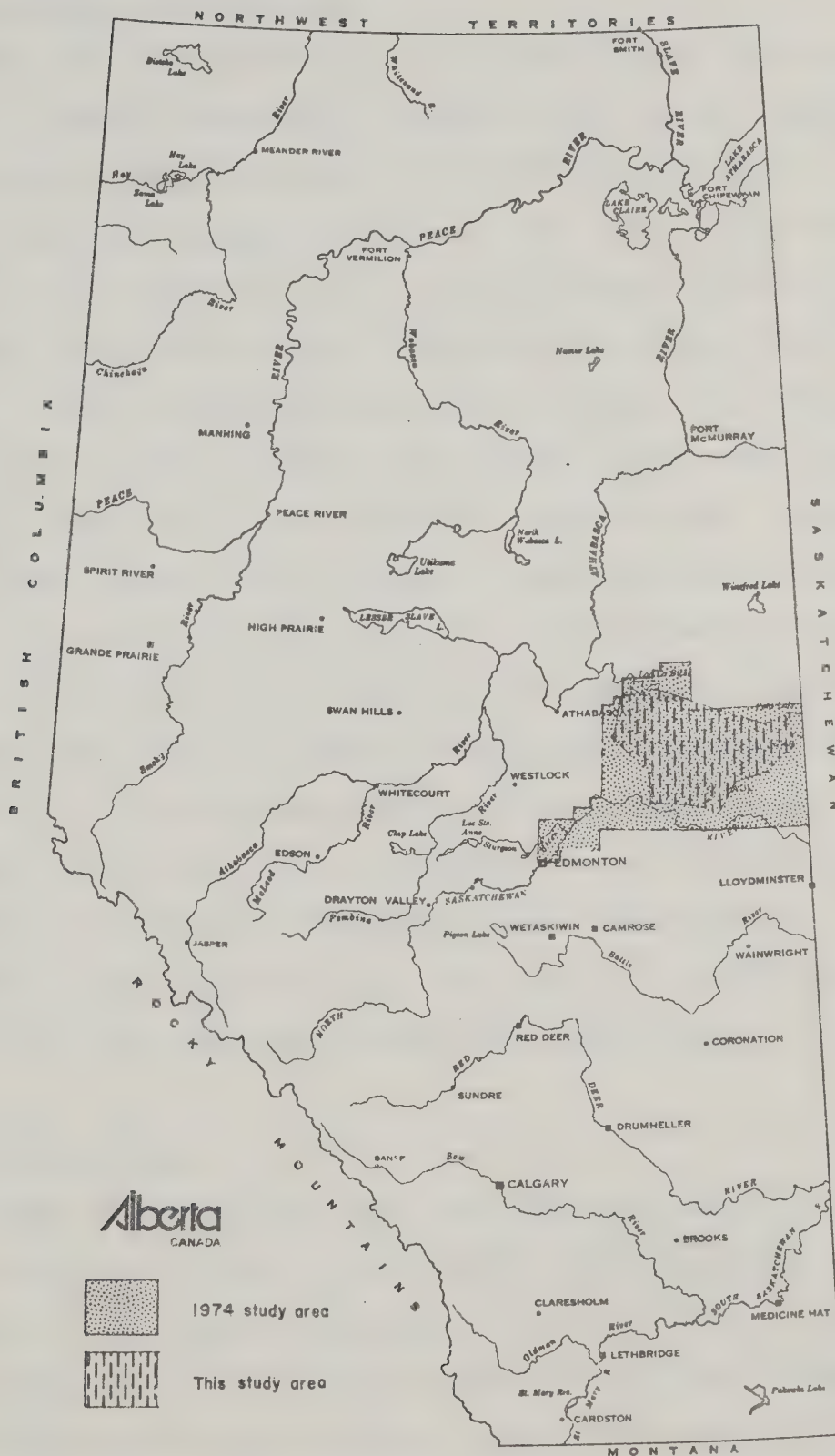


Fig. 1 Index map of the study area.



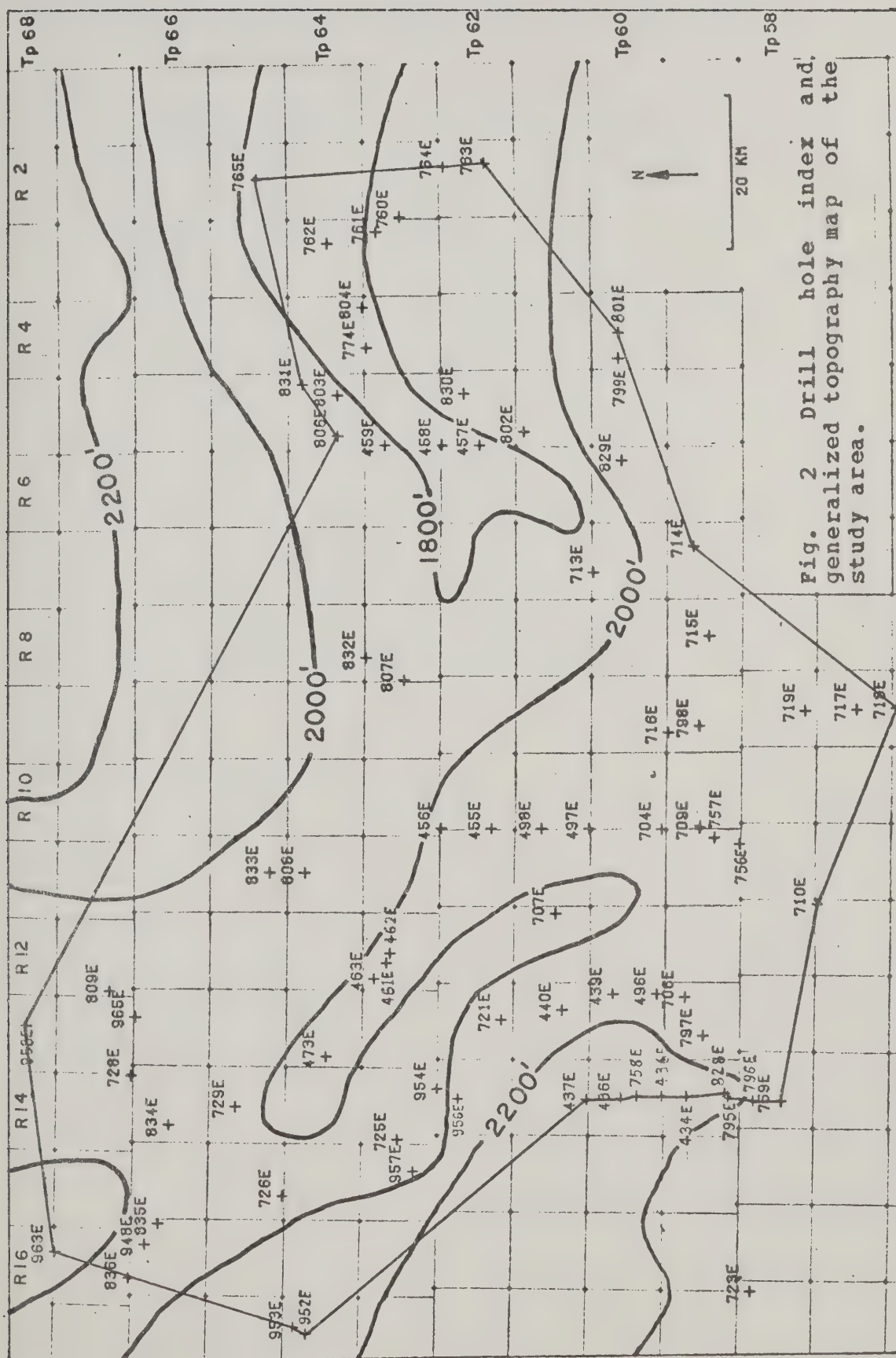
## 2.2 Topography and Drainage

The study area is part of the plains region of east-central Alberta. Topography generally rises to 700 metres in the southwest and slopes gently to the north and east. Moderately higher ground is encountered in the northeast corner of the map area. The low points (below 550 m) in the area are around Lac La Biche and between Moose and Cold Lakes. A broad moderately low region connects them (Fig. 2). Three major drainage systems are present in the area, with little in the way of high ground to separate them. The La Biche River drains the low ground in the northwest and is part of the Athabasca and Mackenzie system, flowing into the Beaufort Sea. The Sand and Beaver Rivers drain to the east via Cold Lake and the Churchill River into Hudson Bay at Churchill. The North Saskatchewan River follows a more southerly route and drains into Hudson Bay at York Factory. The area is covered predominantly with ground moraine (having a local relief of less than 3 m) or hummocky moraine possessing more pronounced mounds and depressions.

## 2.3 Previous Work

Previous work in the area includes the report of Yoon and Vander Pluym (1974) on the buried channels, along with studies of the bedrock topography by Carlson (1967), Lennox and Carlson (1967) and Carlson and Currie (1974). Borneuf (1973) compiled a hydrogeological map of the Tawatinaw map area, which includes the western part of the area of this









study. Le Breton (1963) did the same for east-central Alberta. Bedrock geology has been described by Carrigy (1972) and Green (1972). A hydrogeological map of the map sheet immediately to the east has been produced by Christiansen and Whitaker (1974).



### 3.0 GENERAL LABORATORY METHODS

#### 3.1 Types of Analyses

Four categories of analysis were performed on each of 995 samples from the map area: grain size analysis, lithological analysis by grain counting, carbonate analysis and bulk (major and minor element) chemistry. Complete details of the laboratory procedures used are found in Appendix 1. A general discussion of each is given below along with a statistical summary. Table 1 gives the legal locations of the 78 drill holes and Appendices 2 and 3 give the results of the analyses performed for all 995 samples. Appendix 4 plots these results beside the electric log for each drill hole for ease of examination.

#### 3.2 Sample Examination

The samples used in this project were obtained from holes drilled, logged and sampled by Alberta Environment staff. While it would have been desirable to obtain more frequent and regularly spaced samples from each hole, drilling schedules, the inability to sidewall-sample coarse-grained material and the occasional cave-in prevented this. Because the sampler may not have penetrated the side of the hole precisely where it was intended it was important that each sample was described in the laboratory prior to processing. Comparison with the driller's and electric logs identified any mis-located samples. Because, for example,





Node #	LSD-SEC-TP-RG-W44	Node #	LSD-SEC-TP-RG-W44	Node #	LSD-SEC-TP-RG-W44
432E	LSD 1-28-59-14-W4	717E	LSD 13-15-57-9-W4	953E	LSD 3-14-64-5-W4
435E	LSD 1-4-60-14-W4	718E	LSD 12-34-56-9-W4	954E	LSD 4-1-64-4-W4
436E	LSD 8-21-60-14-W4	719E	LSD 16-4-58-9-W4	806E	LSD 4-17-64-5-W4
437E	LSD 1-4-61-14-W4	721E	LSD 13-3-62-13-W4	807E	LSD 13-18-63-8-W4
438E	SW 25-60-13-W4	723E	LSD 13-25-58-17-W4	808E	LSD 12-27-64-11-W4
440E	SE 15-61-13-W4	725E	LSD 3-19-63-14-W4	809E	LSD 14-7-67-12-W4
455E	LSD 13-7-62-16-W4	726E	LSD 1-5-65-15-W4	828E	LSD 3-10-59-14-W4
456E	LSD 13-31-62-10-W4	728E	LSD 1-1-67-14-W4	870E	LSD 9-23-60-6-W4
457E	NW 18-62-5-W4	749E	LSD 16-21-65-14-W4	870E	LSD 4-36-62-5-W4
458E	LSD 13-31-62-5-W4	756E	LSD 1-2-59-11-W4	871E	LSD 13-25-64-5-W4
459E	LSD 4-30-63-5-W4	757E	LSD 3-13-59-11-W4	872E	LSD 16-32-63-5-W4
461E	NW 23-63-12-W4	758E	LSD 1-16-60-14-W4	873E	LSD 5-10-65-11-W4
462E	NW 21-63-12-W4	759E	LSD 12-16-50-14-W4	874E	LSD 3-20-66-14-W4
483E	NW 31-63-12-W4	760E	LSD 4-15-63-2-W4	875E	LSD 15-24-66-16-W4
473E	NE 13-63-14-W4	761E	LSD 4-36-63-3-W4	876E	LSD 4-4-63-16-W4
475E	LSD 7-1-60-13-W4	762E	LSD 13-14-64-3-W4	948E	LSD 4-35-63-16-W4
477E	LSD 4-6-61-10-W4	763E	LSD 5-14-62-2-W4	952E	LSD 4-27-64-17-W4
480E	LSD 13-19-61-10-W4	764E	NE 34-62-2-W4	953E	LSD 2-34-64-17-W4
704E	LSD 4-6-60-10-W4	765E	LSD 5-15-65-2-W4	954E	LSD 13-34-62-14-W4
706E	LSD 8-26-59-13-W4	774E	LSD 16-32-63-4-W4	956E	LSD 13-15-62-14-W4
707E	LSD 11-13-61-12-W4	795E	LSD 13-3-59-14-W4	957E	LSD 16-10-63-15-W4
709E	LSD 5-19-59-10-W4	796E	LSD 13-28-58-14-W4	958E	LSD 7-15-68-13-W4
710E	LSD 15-36-57-12-W4	797E	LSD 16-17-59-13-W4	963E	LSD 13-35-67-16-W4
713E	LSD 13-33-60-7-W4	798E	LSD 5-21-59-9-W4	965E	LSD 3-2-67-13-W4
714E	LSD 13-23-59-7-W4	799E	LSD 16-19-60-4-W4		
715E	LSD 1-16-59-8-W4	801E	LSD 16-21-60-4-W4		
716E	LSD 15-30-58-9-W4	802E	LSD 5-32-61-5-W4		

Table 1 Legal locations of drill holes used in this study.



some tills, lake deposits and shales may be composed of very similar materials it was of considerable importance that those structural features that could distinguish them were described prior to sample crushing. Only those samples considered to be till were analyzed further.

The term till may have at least two meanings. In the broad sense it includes any material deposited by glacial ice, and in this context extensive thicknesses of bedrock material that have been interpreted as having been moved by the ice have been classified in this study as part of a particular till unit. Dreimanis (1976) discusses deformed bedrock and block inclusions in till. In a more limited sense, till is considered to be a material possessing a characteristically wide range of grain sizes that has been deposited directly from glacial ice. Goldthwait (1971) discusses till as a fairly homogeneous material of mixed particle sizes. It was in this sense that individual drill hole samples were classified as till. While the first of these two properties was readily observable in a small sample, the requirement that it be ice deposited was less readily ascertained than would have been the case for exposed geological sections. Consequently there is a very real probability that some other diamicton was misclassified as till, the most likely candidate being till material that was re-mobilized in some manner, and hence was no longer strictly till. However, this problem has been reduced to some extent by the scale chosen for the current



study. Unlike the examination of a single exposed section, lithologic units much less than 5m thick are often difficult to resolve, due both to the vertical sample spacing and the distance between drill holes. Thus thin diamictons interspersed with water-laid deposits would usually be assigned to a "mixed" category in the log of a drill hole, a not unreasonable classification for a regional-scale study. Thus only extensive thick diamictons would be likely to be stratigraphically considered tills, an interpretation that would probably be correct.

### 3.3 Statistical Evaluation of Laboratory Methods

Laboratory methods were evaluated by analyzing a set of arbitrarily chosen test samples. The results of these analyses are given in Table 2. For the grain size, grain count and carbonate analyses, measurements were taken several times on each of several samples, using different sub-samples of each test sample to produce the replicate analyses of Table 2. In the case of bulk chemistry, an additional step - the preparation of a fused glass of the sample - was needed. Several glasses were made from each test sample and bulk chemical analysis was performed twice on each glass using the electron microprobe. May and Dreimanis (1976) suggest a variability model in which an observation is the sum of the population mean and the natural, sampling, preparation, analytical and random variabilities. Variation between the different analyses for





[illegible]

Table 2 Results of analyses of test samples used to test laboratory procedures for bulk chemistry, grain size, grain lithology and calcium carbonate equivalence.



the bulk chemical analysis may therefore be due to real variation between the samples analyzed (natural plus sampling variability, or between-sample variance in this example), to compositional variation between material taken from the sample to make each glass (preparation variability or sub-sample variance) or to variation between individual measurements on each glass (analytical plus random variation, or error variance). It should be noted that it was not known in advance whether a particular set of samples could be expected to differ significantly in all the analyses performed on it, although a different choice of samples might well have shown such a difference. Thus the ability to distinguish samples in the test set is of less interest than the standard deviation of the within-samples (or between glasses, within-samples) variance. Analysis of variance techniques were used to examine the various components of the variation in laboratory analyses on the test samples. In the case of the bulk chemistry, two arbitrary samples, A and B, each had 4 glasses (sub-samples) made from them. Two analyses were performed on each glass. The analysis of variance model is a one way classification with equal sub-sample numbers (Steel and Torrie, 1960, Section 7.11). The analyses are given in Table 3. The within sample variance may be ascribed to both variation between glasses made from the same sample and variation in multiple analyses of each glass; hence two F tests may be made, comparing variation between samples with variation between



Table 3 Analyses of variance of the bulk chemistry values of Table 2.

Variable	Source of Variation	df	Sum of Square	Mean Square	F	Standard Error of Mean	Standard Deviation	Coefficient of Variation
Na	Between Samples	2	0.025341	0.01267	5.96	.0233	.0659	65%
	Between glasses in samples	2	0.026021	0.00434				
	Between glasses	7	0.051862	0.00741	0.23			
	Within glasses	8	0.154003	0.01925				
	Total	15	0.256551					
Kg	Between Samples	1	0.000676	0.000676	0.18	.0217	.0615	8%
	Between glasses in samples	6	0.027672	0.00461	0.87			
	Between glasses	7	0.023218	0.00332				
	Within glasses	8	0.051673	0.00646				
	Total	15	0.058026					
Al	Between Samples	1	0.013225	0.013225	0.98	.0411	.1163	2%
	Between glasses in samples	6	0.081171	0.01353	1.60			
	Between glasses	7	0.094366	0.01348				
	Within glasses	8	0.067656	0.00846				
	Total	15	0.160018					
Si	Between Samples	1	0.288101	0.288101	3.71	.0965	.2785	1%
	Between glasses in samples	6	0.465334	0.07756	1.60			
	Between glasses	7	0.753434	0.10763				
	Within glasses	8	0.357873	0.04473				
	Total	15	1.464738					
K	Between Samples	1	0.000405	0.000405	0.228	.0165	.0466	2%
	Between glasses in samples	6	0.013018	0.00217	1.26			
	Between glasses	7	0.013513	0.00193				
	Within glasses	8	0.013768	0.00172				
	Total	15	0.027704					
Ca	Between Samples	1	0.001040	0.00104	0.123	.0325	.0920	5%
	Between glasses in samples	6	0.009773	0.00163	0.864			
	Between glasses	7	0.008813	0.00126				
	Within glasses	8	0.008390	0.00105				
	Total	15	0.018003					
Ti	Between Samples	1	0.00063768	0.000638	0.343	.0152	.0431	9%
	Between glasses in samples	6	0.011509	0.00192	1.31			
	Between glasses	7	0.011784	0.00168				
	Within glasses	8	0.0115805	0.00145				
	Total	15	0.023522					
Fe	Between Samples	1	0.124786	0.12479	6.19	.0502	.1420	4%
	Between glasses in samples	6	0.121093	0.02017	1.46			
	Between glasses	7	0.243789	0.03483				
	Within glasses	8	0.082191	0.01027				
	Total	15	0.327959					





glasses, and comparing between glasses and within glasses variance. In the second case significant differences between glasses for the same sample were not found, showing that glass preparation was not a significant source of error.

Table 4 gives the analysis of variance for the grain size, grain count and carbonate analysis. The model is a one-way classification with equal replication (Steel and Torrie, 1960, Section 7.3). Since only one analysis is performed on each sub-sample, the within-sample variance is a combination of sampling and error variance. Duplicate analyses are not performed on each sub-sample in carbonate analysis because the method destroys the material used. For grain counts, analytic error is only found if the operator classifies a grain differently on two different passes, a relatively infrequent occurrence. In addition, the small number of suitable 1-2mm sand grains present in a sample necessitate recombining the analyzed sub-sample with the test sample prior to selecting a further 200 grains for analysis. While it may be possible to perform grain size analysis several times on each sub-sample this was judged to be technically awkward, especially for the silt and clay fractions.

Details of which particular compositional variables differ significantly between test samples are given in the relevant sections below.



Variable	Source of Variation	df	Sum of Square	Mean Square	F	Standard Error of Mean	Standard Deviation	Coefficient of Variation
Size 10-20	Between Samples	4	8.0532	2.0133	7.65	0.363	.513	12%
	Within Samples	5	1.3158	0.2632				
	TOTAL	9	9.3690					
Size	Between Samples	4	10.6669	2.6667	4.27	.559	.790	14%
	Within Samples	5	3.1217	0.6243				
	TOTAL	9	13.7886					
Size 40-60	Between Samples	4	5.2417	2.0604	1.58	.808	1.142	13%
	Within Samples	5	6.5247	1.3049				
	TOTAL	9	11.7664					
Size 60-140	Between Samples	4	65.6494	16.4124	2.18	1.940	2.744	21%
	Within Samples	5	37.6415	7.5283				
	TOTAL	9	103.2909					
Size 140-230	Between Samples	4	10.7552	2.6888	0.76	1.328	1.879	35%
	Within Samples	5	17.6470	3.5294				
	TOTAL	9	28.4022					
Silt Size	Between Samples	4	29.2415	7.3104	1.94	1.371	1.939	4%
	Within Samples	5	18.8267	3.7613				
	TOTAL	9	48.0682					
Clay Size	Between Samples	4	39.8247	9.9562	32.15	.394	.557	5%
	Within Samples	5	1.5484	0.3097				
	TOTAL	9	41.3731					
Local Grains	Between Samples	1	99.225	99.225	11.21	1.330	2.975	21%
	Within Samples	8	70.800	8.850				
	TOTAL	9	170.025					
Acid Grains	Between Samples	1	32.40	32.40	4.531	1.196	2.674	4%
	Within Samples	8	57.20	7.150				
	TOTAL	9	89.60					
Basic Grains	Between Samples	1	8.10	8.10	6.545	.498	1.113	56%
	Within Samples	8	9.90	1.230				
	TOTAL	9	18.00					
Carbonate Grains	Between Samples	1	15.625	15.625	2.372	1.148	2.567	125%
	Within Samples	8	3.60	0.450				
	TOTAL	9	19.225					
Undifferentiated Quarts	Between Samples	1	13.225	13.225	2.713	2.713	2.180	17%
	Within Samples	8	39.00	4.875		.975		
	TOTAL	9	52.225					
Ca CO <sub>3</sub> Equiv.	Between Samples	2	12.7517	6.3759	152.00	.102	.205	4%
	Within Samples	9	0.3775	0.0419				
	TOTAL	11	13.1292					

Table 4 Analyses of variance of the grain size, grain lithology and calcium carbonate equivalence values of Table 2.



### 3.4 Grain Size Analysis

Material from the lightly crushed sample that did not pass through a U.S. standard #230 sieve (0.063 mm) by dry-sieving was washed, dried and sieved through #10 (2 mm), #20 (1 mm), #40 (0.5 mm), #60 (0.25 mm), #140 (0.125 mm) and #230 size sieves. Dry material initially passing through the #230 sieve therefore consisted of silt and clay, and pipette analysis was used to determine the clay content (see Day, 1965). The pipette method was chosen over the hydrometer method due to the reduced labour involved. In the former once particle settling has commenced a single measurement is required after the calculated settling time, whereas in the latter method readings must be taken at systematic intervals, severely limiting the number of samples that may be analyzed at one time. In addition, the pipette method is more readily usable on smaller samples than the hydrometer method. Material between 2mm and 0.063 mm was considered to be sand, between 0.063 and 0.002mm silt and below 0.002mm clay.

In the analysis of variance tests described earlier the test samples could be distinguished by the weight percentages of the 1-2 mm. size fraction at the 95% level, the 0.5 - 1 mm. fraction at the 90% level and the clay fraction at the 95% level. The intermediate size ranges did not differ significantly between the test samples. Although well able to distinguish the test samples, the absolute clay percentages must be suspect due to the long storage times





and consequent disaggregation problems associated with the sidewall samples used (see Appendix 1). Pawluk and Bayrock (1969), for example, found an average of 30% clay (less than 2 micron material) for 20 samples from widely spaced locations in the Sand River map sheet, markedly higher than the 10% observed in the present work. Nevertheless the coarse sand and clay fractions appear to be of use in sample differentiation due to their significant F tests mentioned above. It should be noted that the non-standard granulometric methods described above were implemented due to both the large number and small sizes of the available samples.

### 3.5 Lithological Analysis

Lithological analysis is commonly used in attempts to distinguish between till sheets. Some results from the Edmonton area are quoted in Westgate et al. (1976). His results were obtained by counting boulders in the field or pebbles in the laboratory. Because the size of sidewall samples precluded the use of these size fractions, the 1 - 2 mm sand fraction was used instead. Dreimanis and Vagners (1969, 1971) discuss the use of differing size fractions for examining the variation in lithology in a till, and Willman et al. (1963) use the mineralogy of the sand sizes to distinguish between till sheets. They concluded that sand lithology may successfully distinguish till sheets, but care must be taken not to confuse measurements obtained in one



area using one size fraction with results obtained elsewhere using a different size fraction.

The method used involved the classification of 200 grains from each sample into five categories: local shales and sandstones, acid igneous rocks, basic igneous rocks, carbonate rocks and undifferentiated quartz grains. Five to six samples could be analyzed per hour if interspersed with work less tiring to the eyes. The test samples could be distinguished at the 95% confidence level by the local and carbonate rocks and at the 90% level by the acid and basic igneous rocks (Table 4). As shown by the fairly high coefficient of variation for the basic and carbonate grains, the number of counts observed were rather low for assumptions about the normal distribution of the results to be certain. Statistics based on these two measurements should therefore be treated with caution.

### 3.6 Carbonate Analysis

Among the rock types that were probably traversed by the glacial ice are the limestone and dolomite found in the belt of Devonian rocks fringing the Precambrian shield to the north and east of the map area (Douglas, 1968 and Green, 1972). Rather than counting sand grains to determine the lithology of the parent rocks, however, the content of carbonate rocks is readily determined by mixing the till with hydrochloric acid and determining the amount of carbon dioxide evolved. Dreimanis (1962) and Dreimanis and Vagners



(1969, 1971) used this method to determine calcite and dolomite content of tills in Ontario. Christiansen (1968a,b, 1972a,b) used the same approach to measure the total carbonate content of tills in Saskatchewan. Along with other workers they measured the volume of gas at constant pressure. Dreimanis and Vagners used a manually balanced glass manometer, while Christiansen and Ross (1971) used electro-mechanical equipment of their own design. The advent of relatively inexpensive solid-state pressure sensors permitted the development of a constant-volume device that is inexpensive, reliable and permits rapid analysis even when used by unskilled operators (see Chapter 4). In some parts of North America limestone and dolomite can be distinguished on the basis of their rates of solution, (Dreimanis, 1962). This has also been attempted in Alberta (Westgate, 1972). The current work, however, suggests that this distinction is not easily made and like Christiansen (1972b), only total carbonate or total carbon dioxide evolved should be used, at least in central Alberta. Statistical tests using total carbonate gave excellent sample differentiation in the test analyses. The between-samples variance was significantly greater than the within-samples variance at better than the 99% level (Table 4). Further details of the equipment are given in Chapter 4.





### 3.7 Bulk Chemistry

Bulk chemistry of the silt and clay fraction of till is potentially as useful as carbonate analysis for stratigraphic correlation. May and Dreimanis (1973) have separated tills in southern Ontario on the basis of trace element analysis, and Pawluk and Bayrock (1969) analyzed Alberta tills for some trace elements, but little work has been done on the major and minor rock-forming elements. Energy dispersive analysis on an electron microprobe was used to analyze the till samples for 22 elements simultaneously, 8 of which occurred within the detection limits of the method. Details of the methods developed are given in Chapter 5. Sample fractions passing through a #230 sieve were fused into glasses using the techniques of Schimann and Smith (1976) and analyzed in the microprobe. Raw data were processed using the procedure described in Gold and Smith (1976). With the exception of Na, errors due to preparation of glasses were unimportant on the statistical test data. The samples could be distinguished by Na at the 90% and Fe at the 95% confidence levels, but the Na results were considered suspect because of the high coefficient of variability (Table 3) and the raw analysis values being close to the detection limit for that element. Due to the closeness of their analytic peaks and the low intensities observed it is possible that the Ti figures also include minor amounts of Ba.



## 4.0 CARBONATE APPARATUS

### 4.1 Analytic Methods

The normal method of determining the carbonate content of a soil or till sample requires that the carbon dioxide, generated by reaction with dilute (20%) HCl, be measured by the change in volume at constant (atmospheric) pressure. This is usually done by manually balancing the pressure on either side of a U-tube (Dreimanis, 1962). Other work, attempting to relate the rate of solution to the carbonate composition and grain size, has been done by Skinner and Halstead (1958), Skinner et al. (1959), Turner (1960) and Turner and Skinner (1960). Small negative pressures should always be maintained except when readings are being taken, in order to reduce the chances of leakage. Temperature corrections applied using the gas law,  $PV = nRT$  are usually only applied to the generated gas volume, and ignore the expansion of the dead air in the system, which can cause errors of the same order of magnitude as the measured reading. This is important since the initial heat of reaction is dissipated through the glass walls of the flask in the course of the analysis. Manual reading of volumes is difficult, especially when they must be taken at short time intervals. More elaborate constant-pressure electro-mechanical devices have been developed (Dyck and Perkins, 1974).



#### 4.2 Equipment

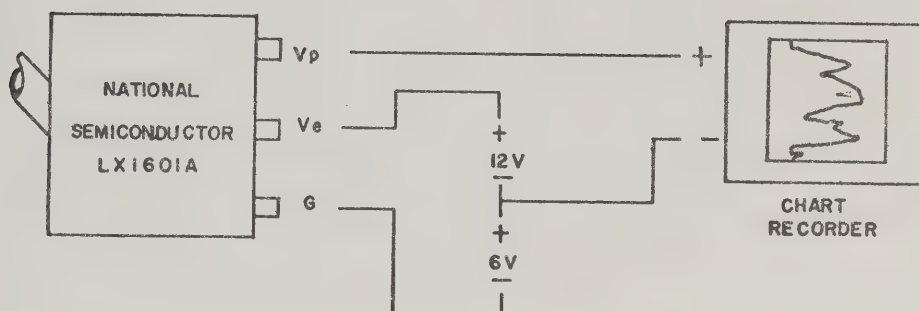
The equipment developed here detects the carbon dioxide evolved by monitoring the pressure change inside a system of constant volume. If pressures are kept low (a few p.s.i.) the effect of pressure variation on the reaction or the solubility of carbon dioxide in the acid will be minimized. The pressure is monitored by a small recently-developed solid-state detector from National Semiconductor and the output fed directly to any chart recorder (Fig. 3a). The detector's sensitivity is much greater than is required for this apparatus. Use of a chart recorder permits use of a stronger acid, usually 50%, and hence a more rapid reaction, easy observation of the end of the reaction and of any errors such as a leak in the flask seal or a slow stirrer. Analysis of typical till samples showed no additional gas evolution for acid concentrations between 20% and 75%, suggesting that for samples used in this study solution of other than carbonate compounds was not a problem. It should be noted that in other areas the use of high acid concentrations may result in the solution of till constituents (e.g. sulphides) not noted in this study.

The detector is connected to an acrylic flask with a screw-down sealing cap and an internal acid reservoir (Fig. 3b). A  $1.00 \pm 0.005$  gm sample and a magnetic stirrer are placed in the flask, the cap screwed down, the flask tilted to mix the acid and sample and then placed on the stirrer. The screw cap minimizes the chances of leakage (which is





### a) ELECTRICAL CIRCUIT FOR CARBONATE APPARATUS



### b) SAMPLE FLASK FOR CARBONATE APPARATUS

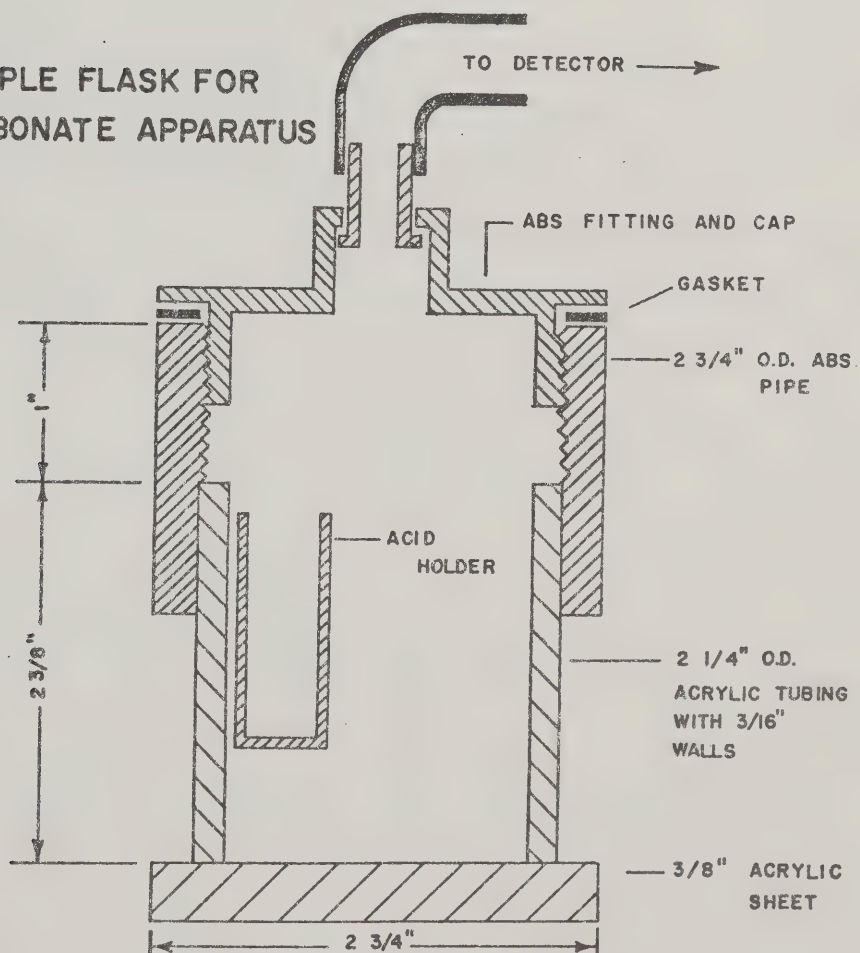


Fig. 3 Construction of carbonate analysis equipment.



rapidly noticed if the cap is not screwed tight enough). With two sets of equipment and spare flasks at least eight samples may be run in an hour, frequently more if the dolomite content is low. Since the equipment is battery operated and portable it may be used in the field if manual agitation and a volt-meter are used instead of the magnetic stirrer and chart recorder, which both require a 110-volt power.

#### 4.3 Temperature Correction

Pyrex glass has a thermal conductivity of 26 calories per second per square cm for a one degree celsius per cm thermal gradient. In the same units acrylic has a thermal conductivity of 4 - 6. Thus as the acrylic flask is a much better insulator than glass the heat of the reaction is retained in the flask with very little loss during the duration of the run. This is supported by the fact that, using the acrylic flask, no pressure drop has been noted for at least 30 minutes after the reaction has ceased. Conversely, with the acrylic flask external temperature changes during a run are of little concern. Magnetic stirrer or hand warmth is capable of penetrating glass, as was readily observed while monitoring pressure changes in a glass flask. For small changes in temperature and pressure the heat of reaction is approximately proportional to the carbonate content of the sample, i.e.

$$[4.1] \quad T_f = T_i + k M_c,$$



where  $M_c$  is the mass of carbon dioxide generated and  $T_f$  and  $T_i$  are the final and initial temperatures of the system. Since the volume  $V$  of the system is constant, as is  $M_a$ , the mass of dead air in the system, the ideal gas law gives the initial pressure of the dead air in the system as

$$[4.2] \quad P_a = R T_i M_a / V,$$

the final partial pressure due to generated carbon dioxide as

$$[4.3] \quad P_c = R T_f M_c / V$$

and the final partial pressure due to the dead air in the system as

$$[4.4] \quad P_e = R T_f M_a / V,$$

$R$  being the gas constant. Substituting for  $T_f$  and simplifying, the final pressure difference

$$[4.5] \quad \begin{aligned} P_d &= P_c + P_e - P_a \\ &= R M_c ( T_i + k ( M_c + M_a ) ) / V. \end{aligned}$$

Thus if the mass of carbon dioxide is a small fraction of the dead air mass the quadratic term can be ignored and pressure change considered proportional to carbon dioxide, and hence carbonate, mass. In most laboratories the dependence on initial atmospheric temperature and pressure can be adequately compensated for by regular calibration with reagent grade calcium carbonate. As shown in Christiansen and Ross (1971) and Perkins and Dyck (1973) it is important that stirring speed be standardized.





#### 4.4 Calibration

Calibration of the apparatus is direct and analyses for calcium carbonate equivalent may be calibrated easily by running a standard of known calcium carbonate composition every few hours, the pressure being nearly proportional to the carbonate content, as shown above. The standard error of analysis is 0.2% (see Table 4), and is partially dependent on the recording device used and the accuracy of sample weighing. The detection limit is visually estimated as 0.2%. Distinction of calcite from dolomite content due to the break in slope of their curve due to their differing solubilities does not appear to be feasible in the test area since the local limestone may take two or three minutes to dissolve (see Fig. 4), as opposed to the less than one minute quoted for limestone in other regions (Dreimanis, 1962). While this may be due to partial dolomitization of the limestone samples, no break in slope was readily apparent in till sample analysis curves. For simplicity of calibration all figures are quoted as calcium carbonate equivalent.



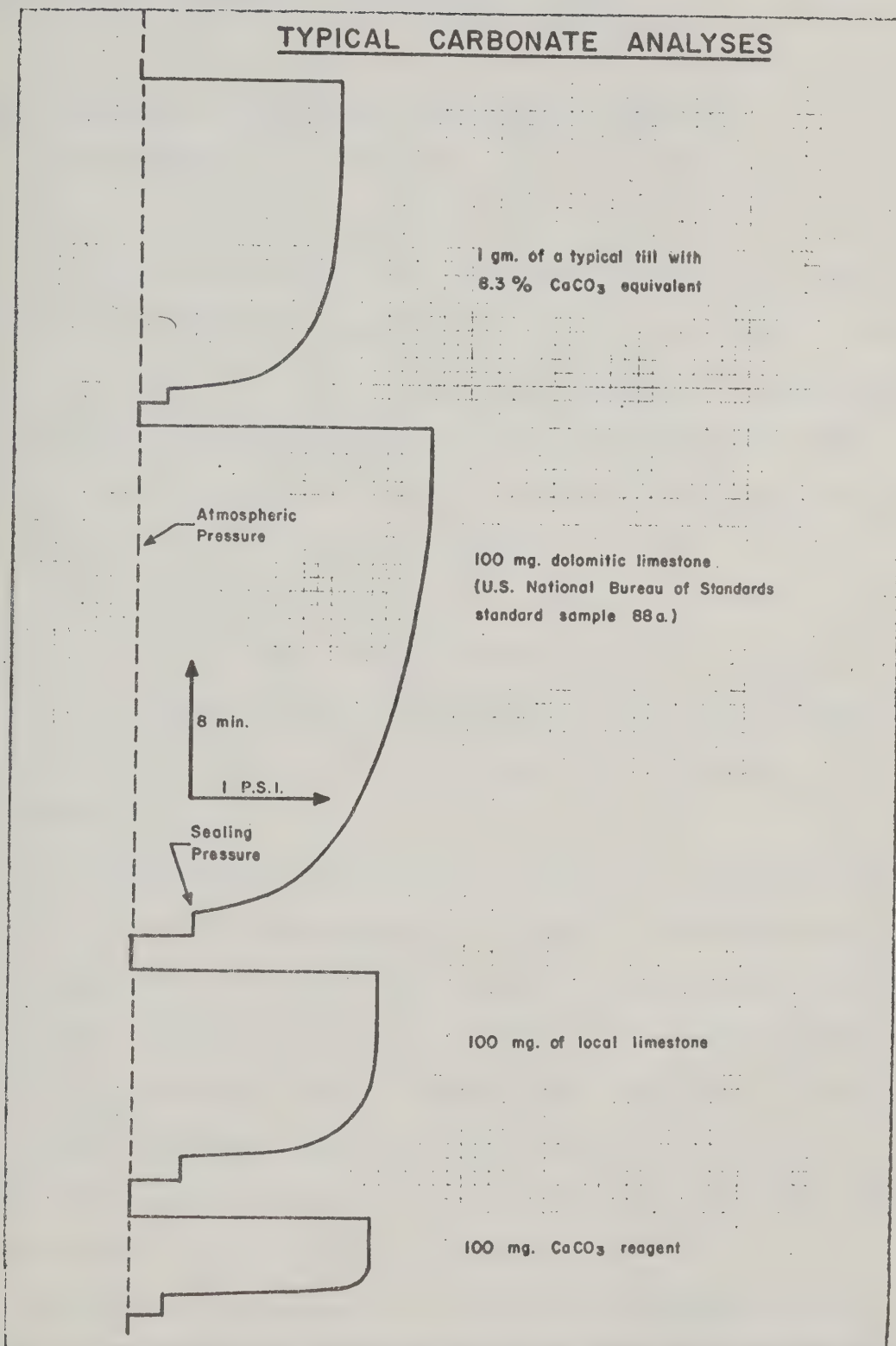


Fig. 4 Plots of pressure (x-axis) versus time (y-axis) for several typical materials, using the apparatus of Fig. 3.



## 5.0 ENERGY DISPERSIVE MICROPROBE ANALYSIS

### 5.1 Principles of Electron Microprobe Analysis

In an electron microprobe, a beam of electrons is focussed on a polished specimen in order to excite the atoms of the sample and produce the characteristic X-ray spectra of elements present. Ignoring various factors such as loss by absorption, the intensity of any particular X-ray line of any element is proportional to the concentration of the element in the sample. Thus if a standard and a specimen are analyzed under the same conditions:

$$[5.1] \quad C_{sp} = C_{st} * I_{sp}/I_{st}$$

where  $C_{sp}$  and  $C_{st}$  are the concentrations in the specimen and standard, respectively, and  $I_{sp}$  and  $I_{st}$  are the measured X-ray line intensities in the specimen and standard, respectively.

The conditions under which a sample may be analyzed, however, can vary appreciably. Different instruments may use different angles of incidence between the electron beam and the sample surface, as well as different take-off angles at which the generated X-rays are sampled. Different operating conditions with the same instrument would include the operating voltage which accelerates the electron beam and the current flowing through the beam.

Since it is necessary to measure the intensity of a particular X-ray line at a particular energy a technique must be used to distinguish between the various lines





emitted. The most common approach is to diffract the X-rays using a suitable crystal and then rotate the detector until the desired wavelength is being sampled. This is described in Rucklidge (1976).

A more recent approach, energy dispersive analysis, uses a solid-state, lithium-drifted silicon ( $\text{Si(Li)}$ ) detector. The size of the pulses output by the detector is dependent on the energy of the impinging X-rays, while the frequency of the pulses depends on X-ray intensity (or rate of photon reception). The pulses are therefore catalogued according to their energy and the counts stored. Thus a typical analysis will result in a spectrum of counts in each of 1024 channels, each 10eV wide. While extremely convenient for simultaneously collecting intensities from several X-ray lines, the ability to resolve different energies is not as good as with diffraction methods. Thus characteristic peaks are relatively broad, causing two particular problems. One is that peaks may overlap each other extensively, making it difficult to estimate the intensities of a particular characteristic X-ray line. The other is that, since peaks are about ten times broader than with diffraction methods, ten times as much of the underlying background intensity is incorporated within any one peak. Attempts to perform quantitative energy-dispersive analysis must include handling of these error sources as well as the usual corrections for absorption and other effects, as described below.



## 5.2 Quantitative Electron Microprobe Analysis

As previously mentioned, the ratio of the intensities of an X-ray line in two samples is approximately the same as the ratios of their concentrations of the element concerned. More accurate analysis, however, requires the introduction of atomic number, absorption and fluorescence corrections. These are discussed in detail in Springer (1976). The atomic number correction includes corrections for the back-scattering of reflected incident electrons, which increases with the average atomic number of the sample. Also included is the stopping-power correction which takes account of the decrease in electrons per unit mass as the atomic number increases. Thus unit mass of elements of low atomic number will absorb a larger proportion of the incident electrons than unit mass of elements of higher atomic number. The proportion of effective (X-ray producing) electrons increases with atomic number due to stopping power (S) but decreases due to backscatter (R) effects.

After X-rays have been generated they may be absorbed by the sample before they leave the surface on their way to the detector. The ratio of transmitted to incident intensity is given by  $\exp(-\mu_m \rho t)$  where  $t$  is the distance travelled,  $\rho$  is the element density and  $\mu_m$  is the mass absorption coefficient for that element for some X-ray energy  $E_\gamma$ . A formula has been derived (Philbert, 1963), making some assumptions about the shape of the volume of sample excited by the electron beam, to express this ratio  $f(X)$  as a



function of  $Z, A, \mu_m, \theta, E_0$  and  $E_c$ .  $Z, A$  and  $\mu_m$  are the atomic number, atomic weight and mass absorption coefficient for the element,  $\theta$ ,  $E_0$  and  $E_c$  are the take-off angle, operating voltage and critical excitation energy for the analysis line.

After generation, some X-rays absorbed by the specimen may have sufficient energy to excite the characteristic radiation of another element present. The X-rays that excite the spurious characteristic radiation may have come from another characteristic line or from the background continuum. The corrections are different in each case and are rather elaborate. Springer (1976) summarizes current atomic number, absorption and fluorescence corrections, and Smith and Gold (1976) describe the equations chosen for correction of energy dispersive analysis data.

### 5.3 Energy Dispersive Correction Procedures

Because many corrections applied to X-ray intensity measurements assume a knowledge of the sample composition, corrections for samples of unknown composition must be applied in an iterative fashion, such that a previous estimate of the composition is used to apply the corrections to the raw intensities, therefore producing a new and, hopefully, better estimate of the composition. When successive estimates agree closely it is assumed that convergence on the correct composition has been achieved. No difficulties have been encountered with this approach if the





initial estimate was reasonable. For energy dispersive analysis in particular, correction procedures may be divided into instrumental calibration corrections, background shaping and scaling, peak overlap correction and composition estimation.

#### 5.4 Instrumental Calibration Corrections

The first requirement for analysis is to know the energy associated with a particular X-ray line intensity. Although an instrument may be designed, for example, to have the lower edge of the first channel set at 0eV and the upper edge at 10eV this will not usually be exact. The approach developed (Smith and Gold 1976) is to collect the spectrum of a calibration standard at the beginning and end of every analytical run. A calibration standard should give strong peaks towards the low and high ends of the spectrum. The common upper limit for analysis being 10 KeV, (10<sup>4</sup> electron volts), willemite was used as the standard, having low energy Zn L<sub>α</sub> and Si K<sub>α</sub> peaks and high energy Zn K<sub>α,β</sub> peaks.

These calibrations permit the estimation of two sets of instrument parameters for the analytic run - those concerning peak position and those estimating instrument resolution. In both cases the first step is the identification of any peaks present and the precise location of their centres. Peak identification is performed by convolution of the spectrum with a 13-point filter as given in Savitzky and Golay (1964, Table III). The filter shown



gives the first derivative of a quadratic function (similar to the top of a Gaussian peak) and hence a change in sign of the derivative from positive to negative will give the peak location to the nearest channel. A closer approximation may be made by making several estimates by interpolation of the full width half maximum (FWHM) of the peak and averaging the mid-points found. The result is usually correct to within 0.1 channels. The FWHM, the usual way of expressing the width of characteristic peaks, is defined as the peak width, measured at a height of half the maximum peak intensity.

The FWHM of a spectrum varies with energy (Smith, 1976). The coefficient  $E_n$  gives the FWHM at zero energy, and the Fano factor  $F$  is a measure of the increase in FWHM with increasing energy. The FWHM at energy  $E$  is  $\sqrt{E_n^2 + 21FE}$ . Hence  $F$  and  $E_n$  for the run may be determined from two peaks in the calibration standard and conversely, once  $E_n$  and  $F$  are known the FWHM is known for a peak at any energy.

If the two energies of the calibration peaks are found, the peaks are identified and the true energies known, it is straightforward to calculate the instrumental stretch (scaling error) and shift (origin offset) for the run and correct all other spectra. The  $F$  and  $E_n$  parameters are used to estimate the FWHM of each analyzed element, and hence the window over which integration of intensity is performed. In this fashion further processing of the spectra is independent of the instrument calibration.

After a spectrum has been corrected for these



instrumental parameters, escape peaks are stripped from it. Escape peaks are produced within the detector itself because there is a small probability that an incoming photon from the sample may excite the Si dead layer on the surface of the detector and that the resulting characteristic Si  $K\alpha$  radiation may re-emerge through the window of the detector and be lost. The correction for this effect is described in Reed and Ware (1972).

Instrumental and escape peak corrections are applied only once to each spectrum. Other corrections, being dependent on the current estimate of the sample composition, are applied once for each iteration.

### 5.5 Background Shape

Ware and Reed (1973) note that the detection limit in energy dispersive analysis is governed more by the limitations in background corrections than by the counting statistics. It is therefore of importance that the background correction should be as accurate as possible, but prior to the current work knowledge of the X-ray continuum shape was incomplete. Ware and Reed (1973) suggested that background intensity  $I$  at energy  $E_{\gamma}$  may be calculated from the expression of Kramers (1923):

$$[5.2] \quad I = kZ(E_0 - E_{\gamma})/E_{\gamma}$$

where  $k$  is a constant,  $Z$  is the average atomic number (proportional to the concentrations of the elements in the sample) and  $E_0$  is the operating voltage used for the





analysis. Additional corrections are made to this theoretical spectrum, primarily for absorption in both the sample and detector. This spectrum is then fitted to the raw sample spectrum by scaling at two positions where peaks are unlikely. Correction for backscatter was not attempted, but Rao-Sahib and Wittry (1972) considered this to be necessary and supplied the relevant coefficients. They also modified Kramers' expression for background intensity to

$$[5.3] \quad I = k(E_0 - E_v)^x Z^n$$

where  $x=1.11$  and  $n$  is a function of  $E_0$ ,  $E_v$  and  $Z$ . However, as noted in Smith et al. (1975)  $I$ , as the number of photons, should be dimensionless, as is Kramers' expression, whereas Rao-Sahib and Wittry's expression includes units of energy. The precise definition of  $n$  was not given, although it was said to vary between 1.2 and 1.37.

Spectra were collected from the following substances: B(metal), C(diamond), MgO(synthetic),  $Al_2O_3$  (synthetic),  $SiO_2$  (quartz), Si(metal),  $TiO_2$  (synthetic), Fe(metal),  $Fe_2O_3$  (hematite),  $Ni_2Si$  (synthetic) and Cu(metal). Initially these were obtained at  $E_0 = 15$  kV (kilovolts). Escape peaks were stripped from the spectra using the method of Reed and Ware (1972). Some of these samples were also run at 5, 10, 12.5, 20, 25 and 29.5 kV operating voltage.

In order to use Kramers' background expression as suggested by Ware and Reed (1973) it is necessary to correct for absorption effects in the detector itself, especially in the low energy end of the spectrum. This requires accurate





knowledge of the thicknesses of the beryllium window, gold contact surface layer and silicon dead layer, with the consequent possibility of error. To avoid this problem Smith (personal communication, 1974) suggested that the ratio, (channel by channel), of two spectra of differing atomic number be used, with adjustments for any differences in probe current or counting time. Thus any deviation of this ratio from unity must be due either to peaks, where it would be ignored, an atomic number-dependent expression for the continuum intensity, or differences due to incorrect absorption corrections. Various attempts were made to see if errors in calculating the atomic number and the absorption matrix corrections could explain the deviations of observed spectral ratios from unity. Mass absorption coefficients were calculated by the formulae of Frazer (1967), Heinrichs (1966) and Kelly (1966). The parameters  $h$  and  $\rho$  used in estimating the actual percentage absorption of a particular wavelength in a particular sample were varied widely. Backscatter corrections were made using the expressions of Duncumb and Reed (1968) and Rao-Sahib and Wittry (1972). None of these adjustments were capable of producing the desired unit ratio between pairs of spectra. Subsequent work was performed using Heinrich's (1966) mass absorption coefficients, Philibert's (1963) absorption corrections formulae and Rao-Sahib and Wittry's (1972) backscatter coefficients. Escape peaks were stripped by the method of Reed and Ware (1972).



Since the ratio variation could not be explained by errors in the matrix corrections, it became necessary to examine continuum intensity expressions. This was done by correcting each spectrum for matrix effects, as described above, dividing it by the proposed continuum expression and then dividing one spectrum by the other. If the sample-dependent component of the expression was adequate the ratio should be unity except for peak regions. For Kramers' continuum expression - that is, using each spectrum divided by its atomic number - the ratios are approximately correct at zero continuum energy but deviate extensively with increasing energy and atomic number ratio. By using ratios of spectra obtained at one operating voltage the terms  $k$  and  $(E_0 - E_\gamma)/E_\gamma$  in Kramers' expression, as well as any detector efficiency terms, are eliminated.

Since Kramers' expression was inadequate and the correct expression clearly involved  $E_0$ ,  $E_\gamma$  and  $Z$  an attempt was made to use Rao-Sahib and Wittry's expression involving  $Z^n$ . Since they stated that  $n$  was dependent on  $Z$ ,  $E_0$  and  $E_\gamma$  but did not give the relationship, a simple additive model was attempted first, i.e.

$$[5.4] \quad n = a + bE_0 + cE_\gamma + dZ.$$

Since ratios are calculated for spectra collected at the same operating voltage  $b$  cannot be determined from a single ratio plot, hence the expression is reduced to

$$[5.5] \quad n = g + cE_\gamma + dZ$$

where  $g = a + bE_0$ . Examination of the ratio at  $E_\gamma = 0$



(obtained by extrapolation) showed this to be slightly greater than unity. A range of values of  $g$  and  $d$  were tried, and the best approximation was found with  $g = 1.06$  and  $d = 0$ . The  $Z$ -dependent term in  $n$  was therefore dropped from further calculations.

If spectra of two substances A and B, obtained at the same operating voltage are compared, the two background shapes can be expressed as  $Z^n$ , and  $n$  is the same for both spectra at any energy  $E_\gamma$ , then the ratio of background intensities

$$[5.6] \quad I_a/I_b = Z_a^n/Z_b^n.$$

Hence it can be shown that

$$[5.7] \quad n = \log(I_a/I_b)/\log(Z_a/Z_b).$$

Fig. 5 shows plots of  $n$  for Cu and C spectra at 5, 10, 15 and 25 kV operating voltage. It can be seen that, allowing for peaks and detector inefficiencies,  $n$  is linear. It should be emphasized that the only assumptions made at this stage are that the continuum intensity may be expressed as a function of  $Z^n$ , and  $n$  itself does not contain a  $Z$  term. Kramers' value of  $n = 1$  plots along the base line, and the ranges of  $n$  found by Rao-Sahib and Wittry (1972) are shown as dashed lines. While  $n$  may fall outside these limits, it does not do so for their combinations of  $E_0$  and  $E_\gamma$ . Since the calculation of  $n$  is very sensitive to intensity fluctuations, care must be taken to obtain spectra at the same operating conditions and with as different atomic numbers as possible for the ratio  $Z_a/Z_b$ . In addition,





regions close to absorption edges should be ignored due to minor errors in the absorption corrections.

The previously-described coefficient  $g$  may be obtained from Fig. 5. Its logarithm, plotted against the logarithm of  $E_0$ , is shown in Fig. 6. Fig. 7 shows the slope of the lines in Fig. 5, plotted against  $\log_{10}(1/E_0)$ . Both plots are approximately linear, and in this fashion errors in obtaining these coefficients graphically may be reduced. When the adjusted lines are plotted on Fig. 5 it is clear that they intercept at a common point ( $N, e$  in Fig. 5). This suggests a further modification to the expression for  $n$ , which may now be rewritten as

$$[5.8] \quad n = N + f(E_0) (E_\gamma/e - 1),$$

where  $f(E_0)$  is the change in  $n$  for  $e$  keV and is a function of  $E_0$ . Fig. 8 shows a plot of  $f(E_0)$  against the logarithm of  $E_0$ , showing good linearity with

$$[5.9] \quad f(E_0) = 0.2532 - 0.0584 \ln(E_0).$$

The resulting expression for  $n$  is

$$[5.10] \quad n = 1.159 + (0.2532 - 0.0584 \ln(E_0)) (E_\gamma/2.044 - 1)$$

which may be simplified to

$$[5.11] \quad n = 1.159 + (0.1239 - 0.02857 \ln(E_0)) (E_\gamma - 2.044).$$

In summary, if characteristic peaks are ignored, a spectrum may be corrected so that its shape is independent of its composition, and hence of its average atomic number  $Z$ . This is achieved by dividing the observed counts by  $Z^n$ , where  $n$  is defined above. If the shape is indeed independent of  $Z$  then the expression  $Z^n$  is a good estimate of the shape



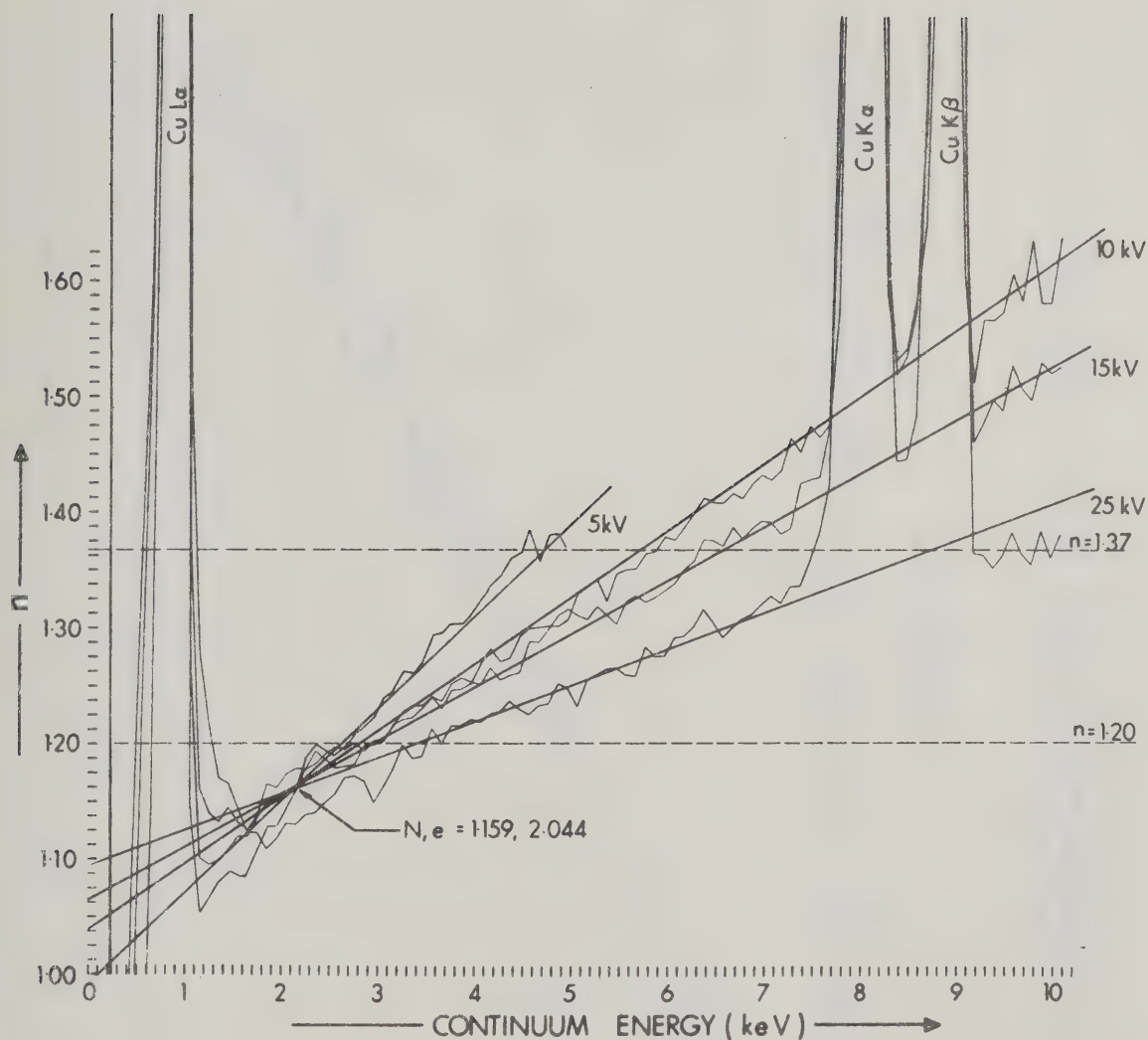


Fig. 5 (From Smith et al., 1975) Plots of  $n$  versus continuum energy for various operating voltages, obtained from the ratios of copper and diamond spectra.



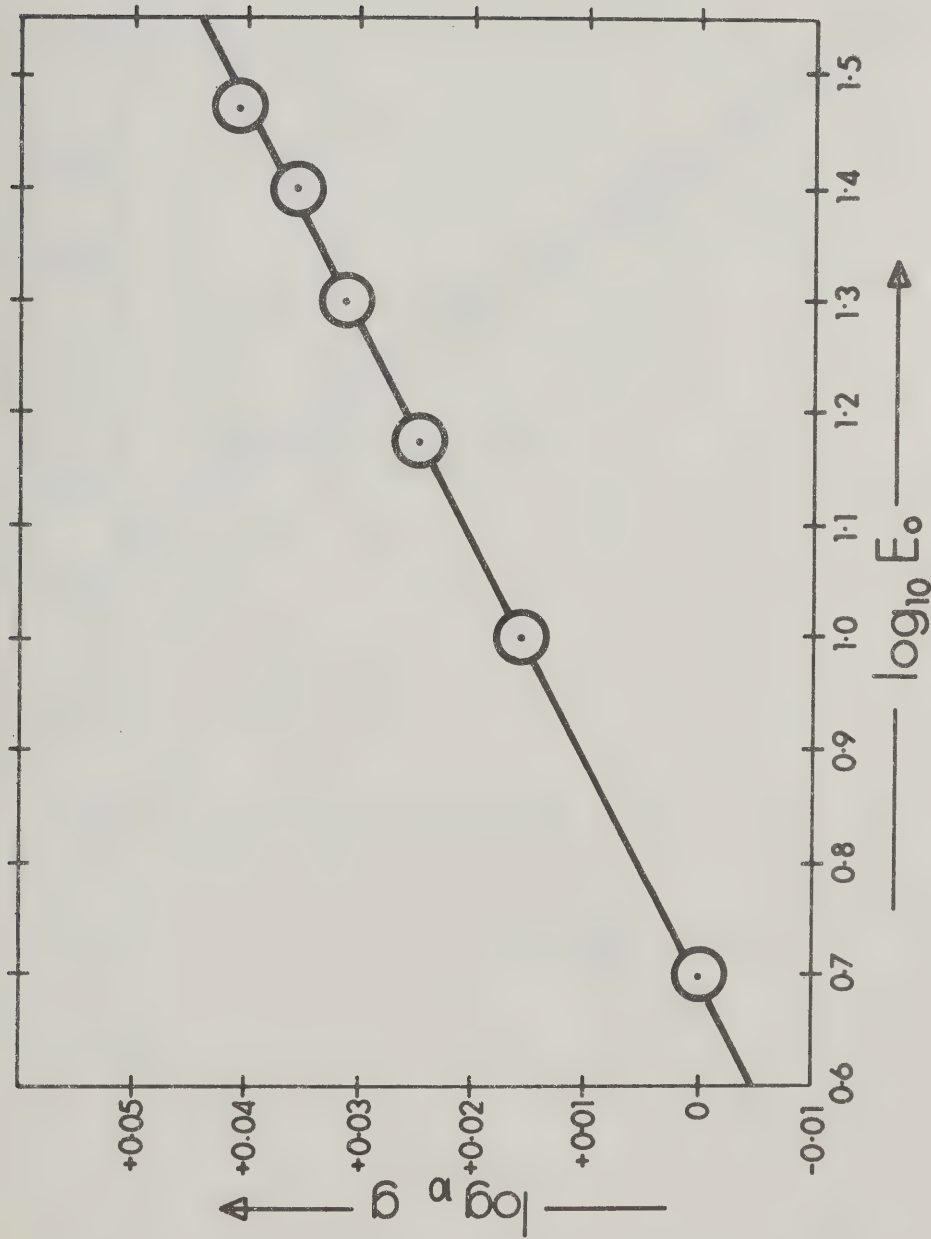


Fig. 6 (From Smith et al., 1975) Logarithmic plot showing the dependence of  $g$  on  $E_0$ .



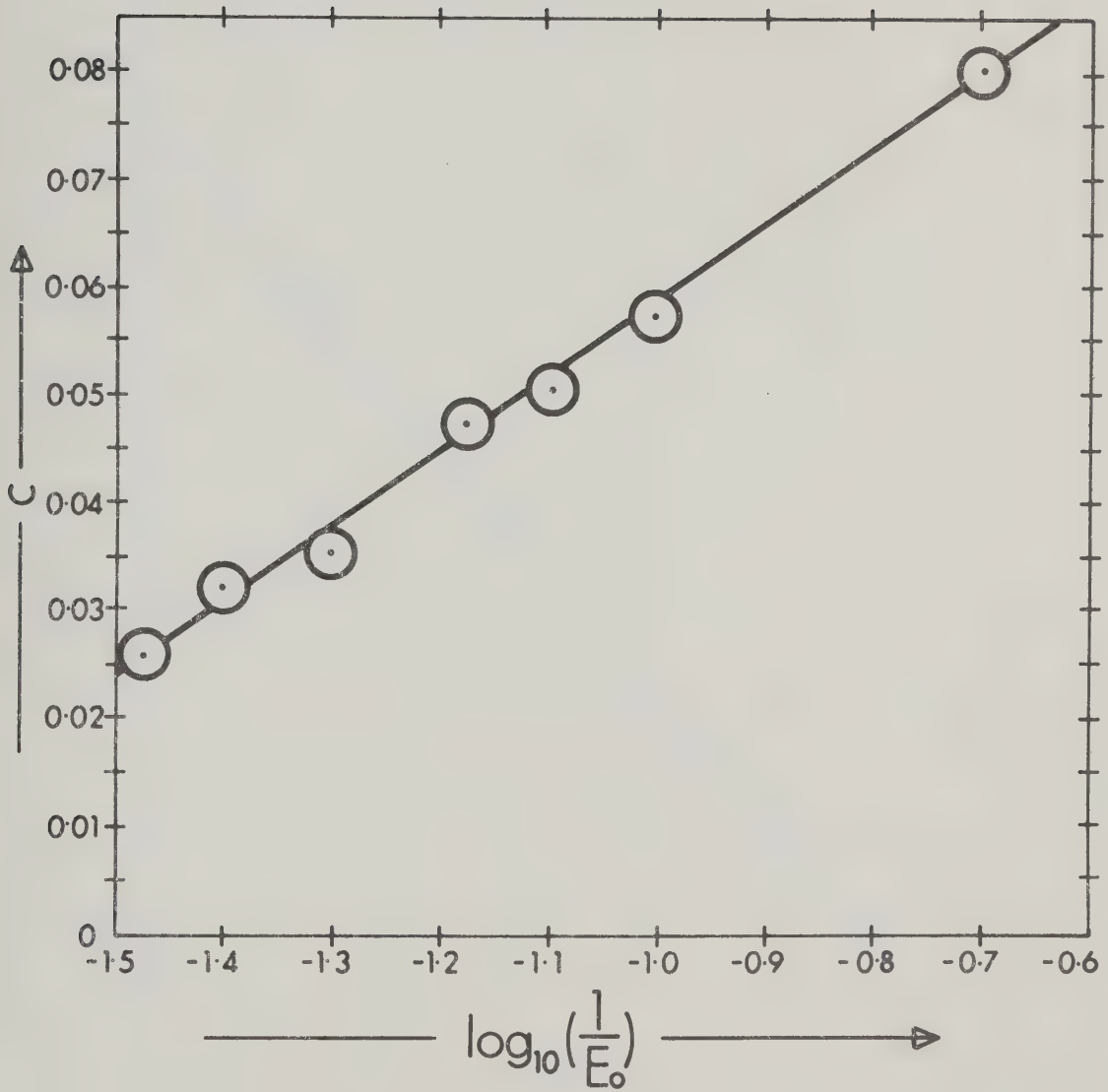


Fig. 7 (From Smith et al., 1975) Plot showing the relationship of  $c$  and  $E_0$ .





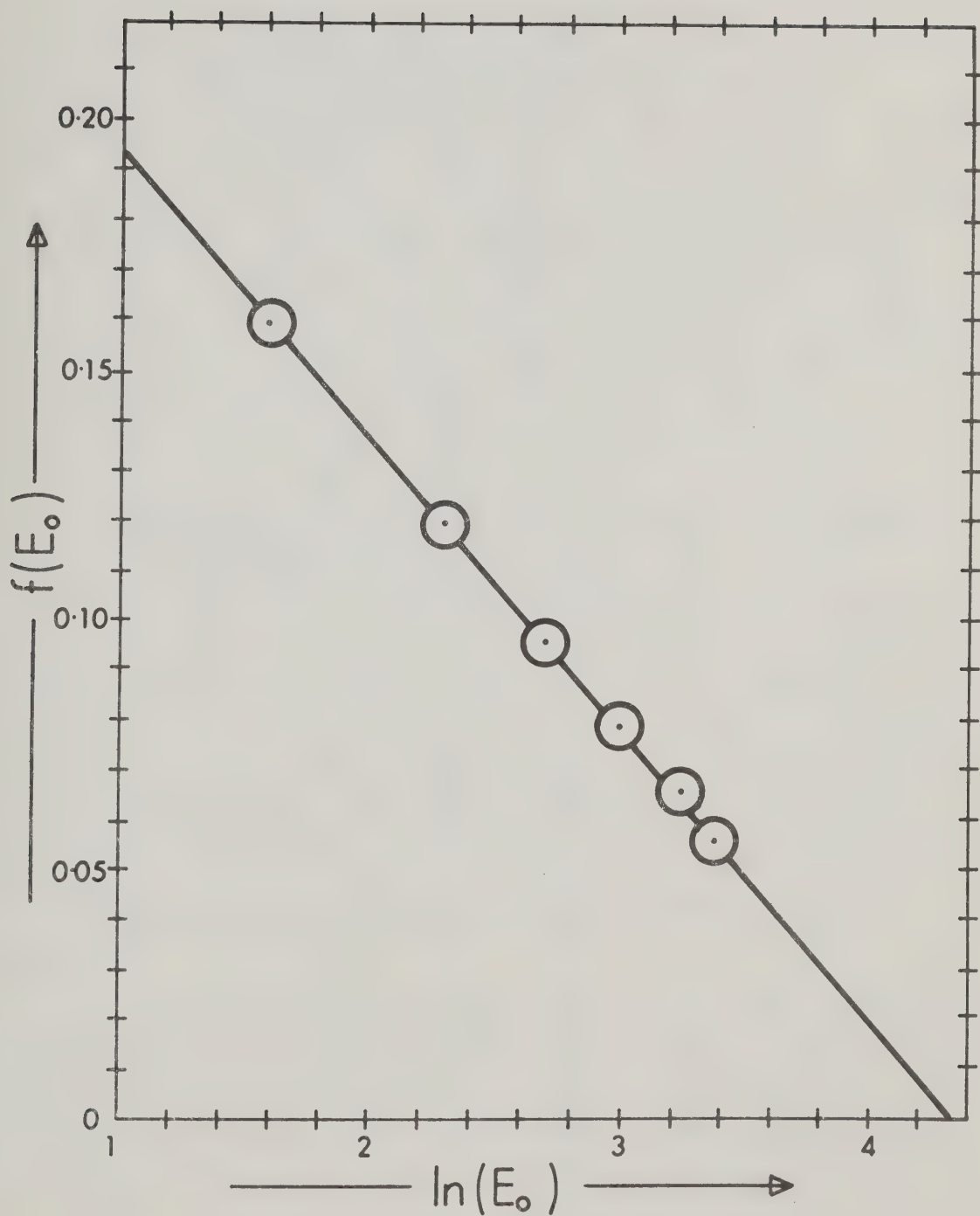


Fig. 8 (From Smith et al., 1975) Plot showing the relationship of  $f(E_o)$  to  $E_o$ .



of the X-ray continuum.

Fig. 9a shows some of the ratios of spectra corrected using  $Z^n$ . Agreement with the expected unit ratio is excellent, as opposed to Fig. 9b which shows similar plots corrected for  $n = 1$  only, as suggested by Kramers. It should be noted that no fundamental significance is necessarily implicit in the form of the expression determined - other expressions may produce even better fits to the data. In addition, the number of significant digits in the coefficients for  $n$  may be fewer than given due to the unknown magnitude of errors in absorption and backscatter corrections and instrumental parameters such as probe current and operating voltage. The new expression does, however, provide a superior estimate of the continuum intensity than previous equations. This is indicated by the close approximation to the unit ratio in Fig. 9a, showing that those factors still affecting the continuum shape are not dependent on the compositional differences between samples.

In order to use the continuum expression directly to simulate a background curve for a sample, it would be necessary to know both the  $E_0$ -dependent term  $-(E_0 - E_v)^x$  in Rao-Sahib and Wittry (1972) - and the detector efficiency curve due to detector absorption of X-radiation. Hence a less error-prone method was implemented instead. Since the operating voltage of the system is not changed frequently, and the detector efficiency should remain constant, a



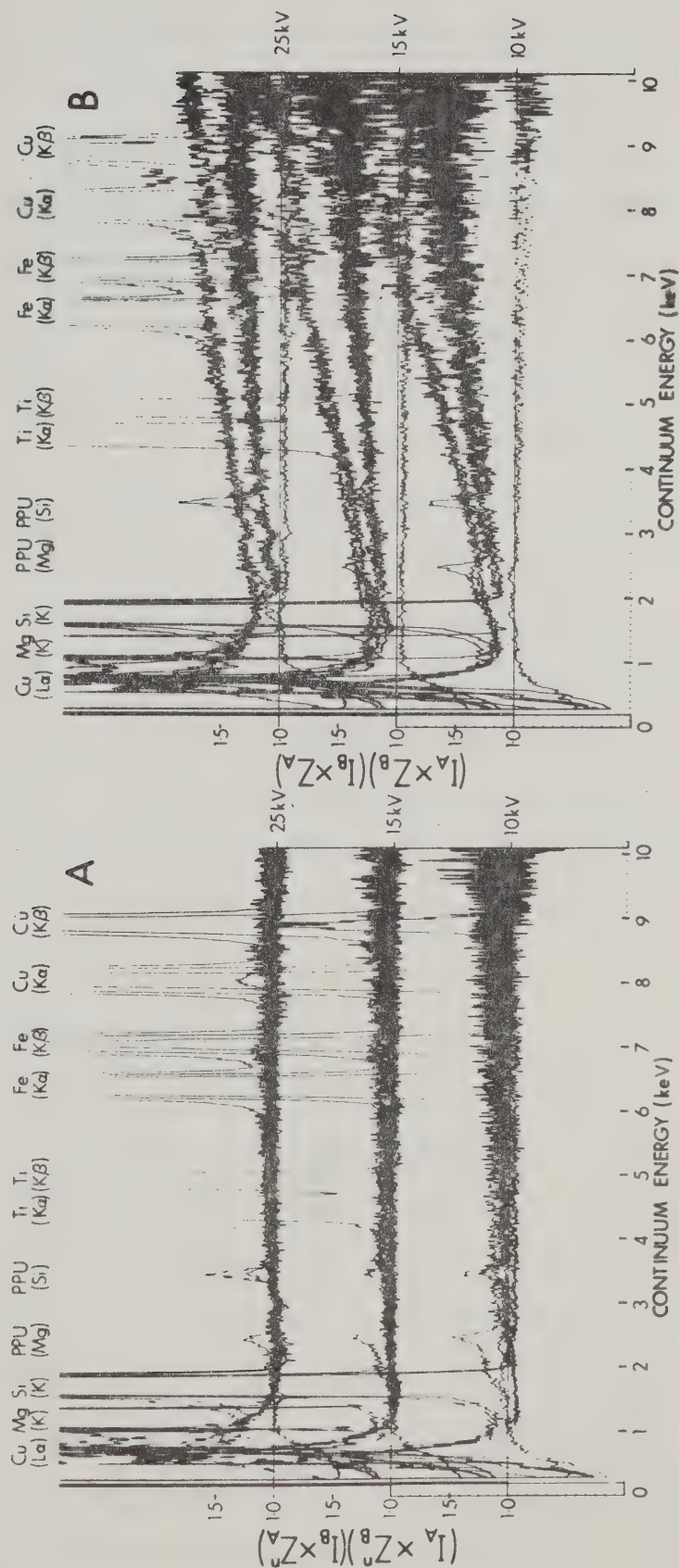


Fig. 9 (From Smith *et al.*, 1975) Plots of the ratios of the Z-corrected spectra of assorted metals and oxides with Z-corrected carbon spectra. Characteristic lines and residual pulse pile-up (PPU) peaks are displayed. (a) shows the approximation of these ratios to unity when the proposed correction is applied. (b) shows the results when Kramers' (n=1) expression is applied.





practical approach is to collect the spectrum of a standard, diamond for example, whose composition is known and which has no characteristic peaks in the region of interest. The spectrum is corrected for matrix effects and divided by  $Z^n$ . This "normalized" background should therefore be free of any composition-dependent effects but should reflect any operating voltage and detector-efficiency characteristics. Once prepared, this normalized background is preserved and used for every specimen analyzed under the conditions at which it was taken. Given the estimated or known sample composition, the background is multiplied by  $Z^n$  and corrected for matrix effects.  $Z^n$  is calculated as  $(\sum_i C_i Z_i)^n$  or, more accurately but with greater computational effort, as  $\sum_i C_i (Z_i^n)$  where  $C_i$  is the concentration of component  $i$ . The resulting spectrum should have the shape of a background produced by a sample of that composition.

### 5.6 Background Fitting

Since different spectra will be acquired with different probe currents and counting times, the shaped background must be scaled to the sample spectrum. Although this could be done using the probe current and counting time, it is more accurate to base the calculations on observed intensities, as this eliminates error due to instrumental drift. Ware and Reed (1973) suggested that scaling be performed using the Ar  $K\alpha$  and Ge  $K\alpha$  energies (2.96 keV and 9.876 keV respectively), which will not normally be



interfered with by other peaks. Disadvantages with this approach are that the Ar  $K\alpha$  position may be interfered with by residual pulse pile-up from Al and the intensity of the Ge  $K\alpha$  position is very low, producing poor counting statistics.

In order to scale the whole spectrum to an overall best fit, it is first necessary to eliminate peaks in any averaging. As a first step  $R_L$ , the lowest spectrum over background ratio (performed using a 13-channel running average to reduce statistical fluctuations) is obtained. If any of the background is clear of peaks, this ratio will approximate the correct scale factor for the shaped background. An upper limit  $R_U$  is then calculated for each channel in turn. A final average ratio is calculated using all channels whose intensity (or number of pulses)  $I$  is less than  $R_U$ , thus eliminating all significant peaks. The upper limit is calculated as

$$[5.12] \quad R_U = R_L (1 + 3.5/\sqrt{I}).$$

The reciprocal of  $\sqrt{I}$  is the standard deviation of the number of pulses  $I$ , if  $I$  is sufficiently large that the binominal distribution of the counts approaches the normal distribution. Thus the upper limit of the acceptable background ratios is set at 3.5 standard deviations above the lowest ratio found in the spectrum. The number 3.5 was found empirically to provide the best discrimination between peaks and background for the instrumentation used. Energies below 2 keV or above 9 keV were not used, due to the steep



slope of the continuum at low energies or the very low count rate at high energies. All spectra were adjusted for stretch and shift, as described earlier in section 5.4, prior to background fitting. Fig. 10 shows the effect of background subtraction on the spectra of ten typical substances. In general the procedure is extremely satisfactory. Slight lows just to the high energy side of some peaks are probably due to inadequacies of absorption corrections for characteristic radiation when applied to the continuum (Smith et al., 1976). Residual counts on the low energy side of peaks just above Si in energy, are low energy tailing and shelving effects due to incomplete charge collection in the silicon detector (Smith 1976).

### 5.7 Peak Overlap

Since energy resolution is much lower in energy dispersive analysis than in wavelength-dispersive analysis, the overlap of peaks is a frequent occurrence. While attempts have been made to distinguish peaks by assuming Gaussian shape, many overlap conditions are not amenable to this approach, particularly where the non-Gaussian tails of peaks are involved. Consequently a more general approach was applied.

Peaks are separated so that a central part of each may be integrated to produce a measure of the total intensity of that X-ray line. Since  $F$  and  $E_n$  are estimated for the run, and spectra are therefore automatically corrected for





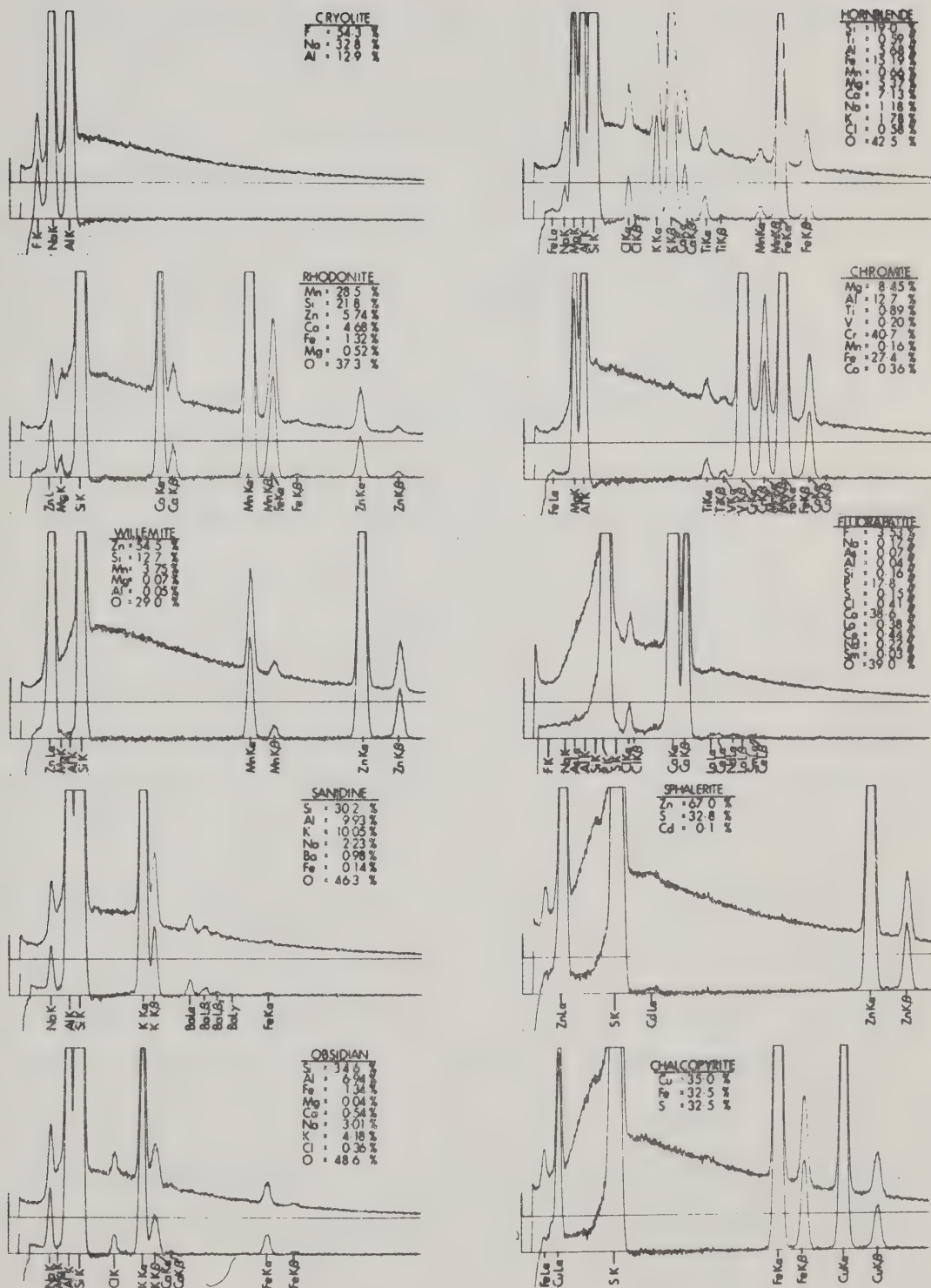


Fig. 10 (From Smith et al., 1975) Spectra of various compounds before (upper) and after (lower) continuum and escape peak removal.





stretch and shift, it is possible to define a run-condition dependent set of integrating regions each covering 1.2 full width half maxima around the peak centre for each analytic line. It is therefore possible to process a set of simple elements or compounds and obtain the above-background intensities that fall within the defined integrating regions. These are expressed as a fraction of the intensity obtained for the analytic line of the element being processed.

Table 5 shows the overlap coefficients for 22 elements commonly used in analysis of silicate minerals. Fig. 11 shows the main features of this table diagrammatically. Overlap coefficients decrease progressively on the low-energy side of parent peaks, but increase as the energy of the parent peak approaches that of the silicon absorption edge. "This pattern indicates that the shelving effects reflected in these overlap coefficients, result largely from partial loss of charge in the silicon deadlayer, as well as from trapping in impurity centres in the detector." (Smith and Gold, 1976, p. 196). In addition, where overlap is produced by element peaks other than the analytic line, differential absorption corrections should be applied. This is the case for the  $K\beta$  peaks on the high energy side of the analytic  $K\alpha$  line. In addition, atomic number and fluorescence corrections are applicable where  $L\alpha$  peaks overlap some regions of interest and analytic results are obtained using the  $K\alpha$  peak for that element.



F	Na	Mg	Al	Si	P	S	Cl	K	Ca	Sc	Ti	V	Cr	Mn	Fe	Co	Ni	Cu	Zn	Zr	Ba
100	2.10																				
2.01	100	0.70																			
0.55	1.62	100	0.21																		
0.35	0.46	1.29	100	0.12																	
0.29	0.23	0.30	0.90	100	0.19																
0.96	1.12	1.55	2.15	4.13	100	0.70															
0.70	0.67	0.77	0.99	1.40	2.81	100	1.16														
0.55	0.47	0.43	0.46	0.59	0.90	1.96	100	0.10													
0.49	0.44	0.32	0.22	0.11	0.14	0.24	0.34	100	7.16												
0.45	0.35	0.26	0.12	0.08	0.12	0.19	0.33	0.63	100	10.03											
0.38	0.31	0.20	0.06	0.02	0.03	0.07	0.13	0.23	0.51	100	12.35										
0.34	0.28	0.17	0.03			0.02	0.05	0.10	0.20	0.40	100	13.70									
0.30	0.24	0.14	0.01				0.01	0.04	0.08	0.17	0.32	100	12.65								
0.25	0.20	0.10	0.01						0.03	0.07	0.15	0.26	100	11.10							
0.22	0.16	0.07	0.02							0.02	0.05	0.13	0.21	100	8.20						
1.29	1.50	0.23	0.03								0.02	0.06	0.13	0.18	100	5.00					
1.26	1.10	0.30	0.05								0.02	0.06	0.06	0.12	0.15	100	2.50				
1.24	0.89	0.39	0.05									0.02	0.01	0.05	0.09	0.13	100	1.30			
1.18	24.51	0.61	0.10												0.03	0.06	0.10	100	0.71		
4.26	282.1	2.07	1.57														0.02	0.07	100		
0.69	1.14	1.59	2.08	4.69	80.43	5.05	0.13													100	
1.08	1.92	0.88	0.39	0.25	0.25			0.07	0.33	2.03	95.63	33.38	2.99	2.19						3.11	100

Overlap coefficients expressed as percentages of measured intensities (1.2 FWHM) in analysis peaks of overlapping elements. N.B.: where L line overlap coefficients are expressed as percentages of associated K lines, full differential ZAF corrections are applicable. Coefficients in this table are applicable only at the operating voltage used (15 kV) and, strictly, only at the average counting rate of 3000/sec (full spectrum) at which they were obtained. Changes in counting rate producing significant peak broadening will change the coefficients. Also, they are applicable only to a particular detector resolution.

Table 5 (From Smith and Gold, 1976)



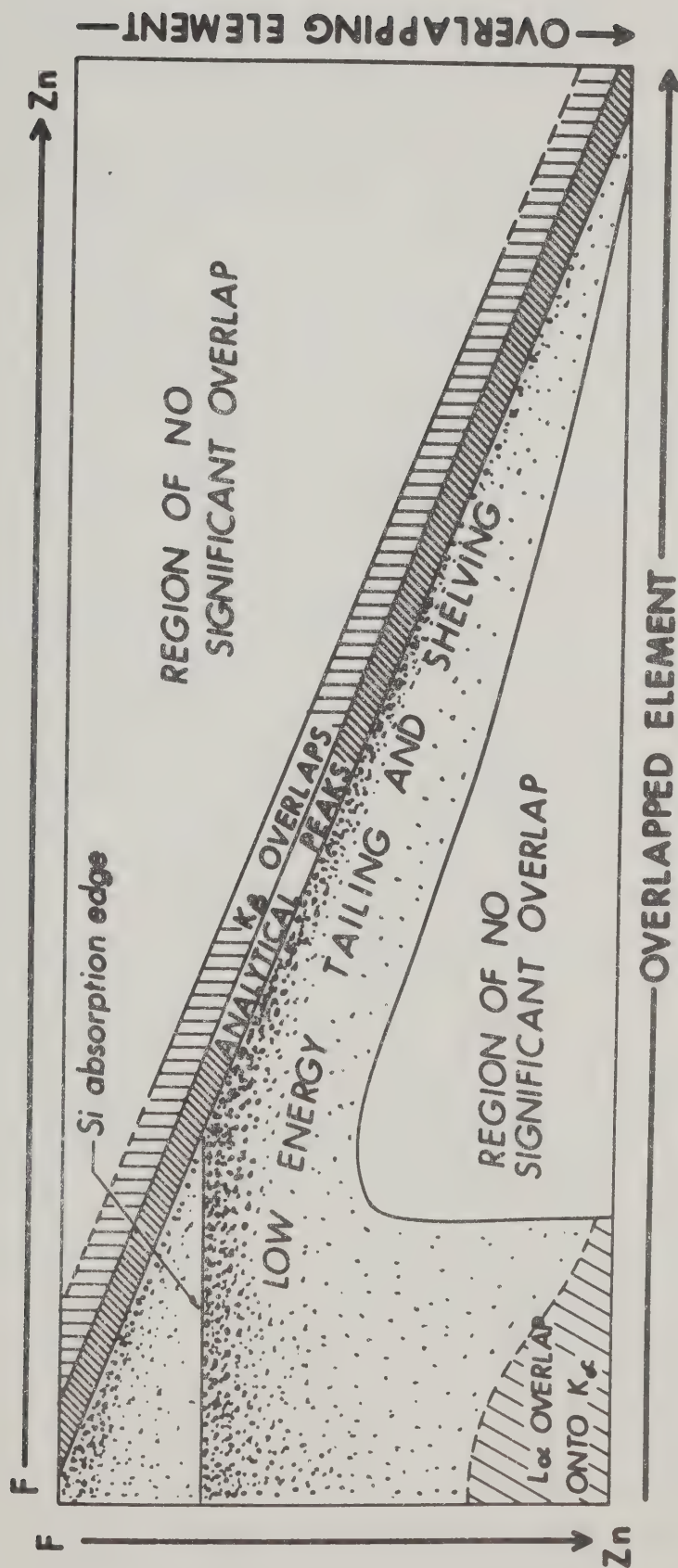


Fig. 11 (From Smith and Gold, 1976) Diagrammatic representation of the overlap coefficients from F to Zn of Table 5.





Since the measured intensities for a sample are the end result of all relevant overlaps, and the overlap coefficients are known, the solution of a set of 22 simultaneous equations for the 22 intensities measured at the locations of the analytic peaks provides the solution for the 22 original intensities prior to overlap. Overlap coefficients are, however, affected by the system resolution, which is affected by the counting rate. Hence all spectra should be obtained at similar counting rates.

### 5.8 Calculation of Composition

Once intensities are obtained they are corrected for matrix effects as described in section 5.2. Corrected intensities obtained from a sample of unknown composition must then be compared with intensities obtained from standards of known composition in order to estimate the concentration of any particular element. Thus

$$[5.13] \quad C_{sp}/I_{sp} = C_{st}/I_{st}$$

where sp and st refer to specimen and standard, respectively, and C and I are concentration and corrected intensity, respectively. Standards must therefore be processed in each run for each element analyzed, prior to the analysis of an unknown. Fig. 12 shows a typical printout for a standard, using the EDATA program (Gold and Smith, 1976). While this is undoubtedly the most accurate method, a considerable saving in analysis time can be made if approximate standard intensities can be preserved from



```

*****
NEW SPECTRUM, TIME= 400 SEC.

COMM0279 SANIDINE SMM4-21
PERCENTAGES:
  30.23SI 14 0.0 TI 22 9.93AL 13 0.14FE 26 0.0 CA 20
  2.23NA 11 10.05K 19 0.98BA 56 46.28D 8
AV. Z= 11.9847
MIN E = 208
SBR= 0.003022

PEAK INT. P/B R. FWHM.
1484.01 425362 8.56 126.05 AL KA
1736.19 1538250 26.97 126.10 SI KA
3313.64 416700 10.76 141.01 K KA
4459.83 11866 0.36 145.30 BA LA LOW P/B
CALIBRATION STANDARD: SHIFT= 1.933 STRETCH=-.00044 USED.
FAND= 0.114934 ENDISE=110.363052 USED
MIN E = 208
SBR= 0.998232
OVERLAP CONVERGENCE: ITERATIONS= 25

Z R F ZAF CTS C/S ZAF+C/S STD. WT.FR. EL.
1.0794 1.0576 0.9996 1.1410 425832. 1064.58 1214.71 0.0000827 0.10050 K
1.5529 0.9877 0.9998 1.5335 11156. 27.89 42.77 0.0002291 0.00980 BA

```

Fig. 12 Sample EDATA printout for a standard.



previous runs for use in estimating the concentrations of minor or unexpected elements. For this purpose standard intensities are preserved in a standardized form suitable for use in any run. Intensities are recorded as counts per second per nanoamp of probe current per percent of that element present in the standard. While not as accurate as run-time standards they have appreciable value. Fig. 13 shows the EDATA printout for a typical sample analysis. In the last line under "OLD STDS.", can be seen the standard numbers used for each element. In this list a zero indicates that default standard values were used.

#### 5.9 Sequence and Implementation of Correction Procedures

The normal procedure for analysis of an unknown therefore contains the following steps.

1/

The sample spectrum is corrected for stretch and shift and escape peaks are stripped from it.

2/

The spectrum is scaled to standard probe current.

3/

An initial estimate of composition is made.

4/

The normalized background is shaped using the current composition estimate, involving the application of matrix and  $Z^n$  corrections.

5/



Fig. 13 Sample EDATA printout for a sample of unknown composition.





The background spectrum is scaled to fit the sample spectrum, and subtracted from it.

6/

Integrated intensities are obtained over standardized regions of interest for each element, using the background-stripped sample spectrum.

7/

The integrated intensities are corrected for peak overlap.

8/

Resulting intensities are corrected for atomic number, absorption and fluorescence effects.

9/

New compositional estimates are made by comparison with standard intensities.

10/

If the current and previous composition estimates are sufficiently close (within 0.05% for all elements) calculation ceases. Otherwise the procedure is repeated from step 4.

Fig. 14 shows a schematic flow chart of the EDATA package. After subroutine INIT initializes parameters for the run a spectrum is read and standardized by subroutine READIN, and, if this spectrum is of a sample of unknown composition, the initial composition is estimated from the raw peak intensities by calling SCAN with a parameter of zero. After this, subroutine SHAPE is called to shape the



# EDATA

## SCHEMATIC FLOW CHART

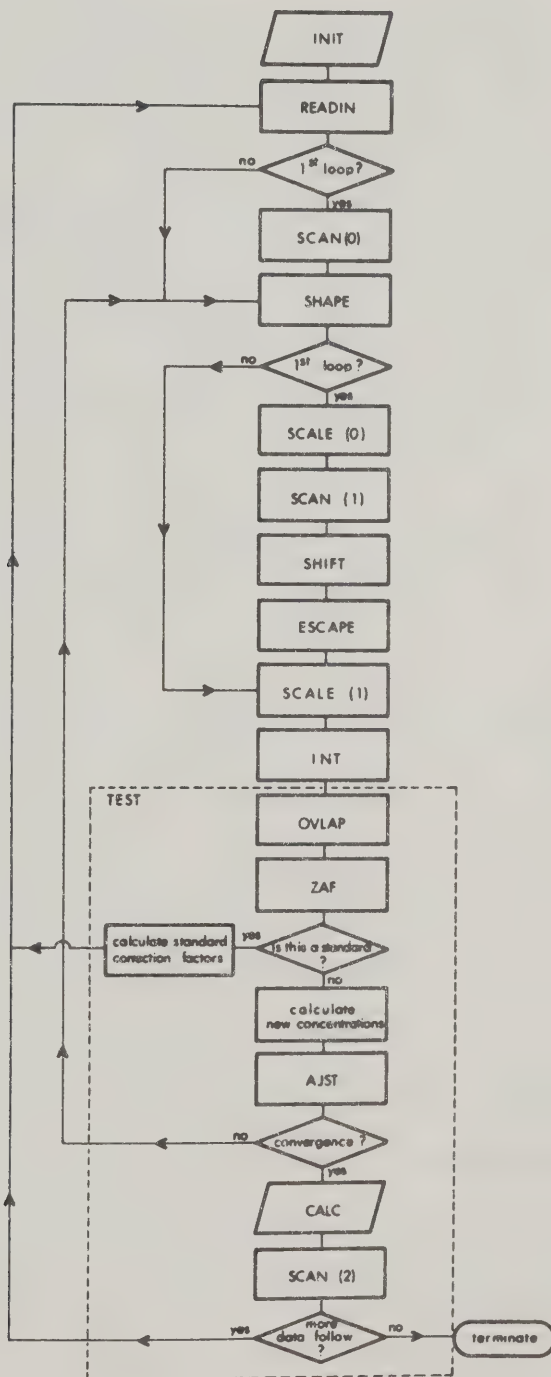


Fig. 14 Schematic flowchart for the EDATA program.



normalized background to the sample spectrum, SCALE with a parameter of zero is used to fit the shaped background to the raw spectrum, and SCAN with a parameter of one is used to search for and display all major peaks and their size with respect to the background (see Figs. 12 and 13). Following this, SHIFT is called to adjust the raw spectrum for any instrumental stretch and shift as estimated from the calibration standards, ESCAPE is called to strip off the escape peaks, and SCALE with a parameter of one does a final scaling adjustment to fit the background to the shifted sample spectrum. Subroutine INT then integrates both the shifted spectrum and the scaled background over the region of interest for each of the 22 elements, as estimated from the FWHM of the calibration standards. The spectrum-minus-background figure for each element forms the intensity value used to generate subsequent compositional estimates.

Subroutine TEST handles all subsequent processing of intensities. OVLAP is called to estimate the original intensities in the absence of any peak-to-peak overlapping. ZAF corrects these intensities for atomic number, absorption and fluorescence effects. Processing of standards terminates at this point with the derivation of standard correction factors in the form of weight fraction of an element per ZAF-corrected count / second intensity, to be used for compositional estimates on subsequent samples. Thus, in Fig. 12, the values in the "STD." column are obtained by dividing the known standard composition values in the "WT.FR." column





by the intensities in the "ZAF\*C/S" column. In the case of a sample being analyzed (Fig. 13) the new concentrations ("WT.FR." column) are calculated by multiplying the intensities in the "ZAF\*C/S" column by the values under "STD.". Subsequently a user-defined AJST routine is called to modify the concentration estimates, e.g. by estimating the oxygen concentration as the difference between the compositional total and 100%. This compositional estimate is then used in loop 2. On the second or subsequent iteration only SHAPE (to reshape the background on the basis of the new composition estimate), SCALE with a parameter of one (to scale this background to the initial shift, stretch and escape peak corrected spectrum) and INT (to calculate peak intensities) are called prior to re-entering the TEST subroutine. After new concentration estimates are obtained they are compared with the previous values, and a further iteration is performed if any element deviates from the previous value by 0.05% or more. If, however, convergence has been achieved, the user-defined subroutine CALC is entered (perhaps to recalculate elemental concentrations as oxides), SCAN with a parameter of two searches for peaks on the background-subtracted spectrum (Fig. 13, bottom) and the program proceeds to read and process any subsequent spectra.

Further details of the computer implementation of this approach as the EDATA package are contained in Gold and Smith (1976). Typical analyses obtained using this system are shown in Table 6, from Smith and Gold (1976). For major



	FORSTERITE	SANIDINE	ANHYDRITE	SPHALERITE	TROILITE	OBSIDIAN	GLASS WRAB-4	GLASS BCR-1	ILMENITE		
O	45.58 45.48	45.92 46.28	46.06 47.01	-	-	49.13 48.63	44.52 45.01	44.69 44.73	31.40	31.72	30.57
Na	-	2.10 2.23	-	-	-	3.04 3.01	1.89 1.70	2.58 2.50	-	-	-
Mg	34.29 34.56	-	-	-	-	- 0.04	6.15 5.68	1.97 2.10	0.43	0.58	0.41
Al	-	-	-	-	-	6.88 6.94	9.52 9.56	7.12 7.39	-	-	-
Si	19.97 19.96	10.22 9.93	-	-	-	34.58 34.56	22.48 22.57	25.71 25.72	-	-	-
P	-	30.22 30.23	-	-	-	-	-	0.14 0.20	-	-	-
S	-	-	23.62 23.55	33.32 33.10	36.48 36.50	-	-	- 0.03	-	-	-
Cl	-	-	-	-	-	0.30 0.36	-	-	-	-	-
K	-	10.15 10.05	-	0.06	-	4.20 4.18	0.05 0.13	1.39 1.42	-	-	-
Ca	-	0.23 0.01	30.27 29.44	-	-	0.61 0.54	8.03 8.23	4.95 5.04	-	-	-
Sc	-	-	-	-	-	-	-	-	-	-	-
Ti	-	-	-	-	-	-	0.44 0.41	1.42 1.36	28.78 28.59	29.78	28.78
V	-	-	-	-	-	-	-	- 0.04	0.37	0.17	0.37
Cr	-	-	-	-	0.06 0.08	-	-	-	-	-	-
Mn	-	-	-	-	-	-	0.12 0.14	0.13 0.14	0.32 0.23	0.30	0.30
Fe	-	0.11 0.14	-	2.68 2.90	63.05 63.16	1.18 1.34	6.77 6.69	9.67 9.60	38.67 38.69	38.58	38.58
Co	-	-	-	-	0.24 0.05	-	0.04	0.05	-	-	-
Ni	0.05	-	-	-	- 0.10	-	-	-	-	-	-
Cu	-	-	-	-	-	-	-	-	-	-	-
Zn	0.05	-	0.05	64.78 64.10	0.45	0.07	-	0.08 0.01	0.03 0.01	-	-
Zr	-	-	-	-	-	-	-	-	-	-	-
Ba	0.06	1.01 0.98	-	-	-	-	-	0.18 0.08	-	-	-

Typical analyses obtained using techniques outlined.

The first column of each pair shows results of energy dispersive analysis; values not underlined were obtained by the default standard option; all oxygen values are calculated by difference from 100%. The second columns show the best available information on composition. The third column of the ilmenite analysis shows values obtained using the default option for all elements, including majors.

Table 6 (From Smith and Gold, 1976)



elements the accuracy is possibly better than that obtainable by wavelength dispersive analysis, since all elements were analyzed at exactly the same points on the sample, and their intensities were collected simultaneously. For minor elements, present in concentrations greater than about 0.25%, accuracies are similar, but below this accuracies become appreciably worse than with wavelength dispersive analysis, although with care concentrations of 500 ppm may be detected. Currently the detection limit is considered to be 300ppm. In Table 7 the best available analyses found in the literature (column 2) for an obsidian sample are compared with an analysis using energy dispersive analysis and EDATA (column 1). Elements underlined were calculated using standards processed during the run, while Cl and Mn were calculated using default standards.



OBSIDIAN

	(1)	(2)
O	48.61	48.63
Na	<u>3.19</u>	3.01
Mg	-	0.04
Al	<u>6.91</u>	6.94
Si	<u>34.94</u>	34.56
P	-	-
S	-	-
Cl	0.32	0.36
K	<u>4.36</u>	4.18
Ca	<u>0.52</u>	0.54
Sc	-	-
Ti	-	0.04
V	-	-
Cr	-	-
Mn	0.05	0.05
Fe	<u>1.15</u>	1.34
Co	-	-
Ni	-	-
Cu	-	-
Zn	-	-
Zr	-	-
Ba	-	-

Table 7 Comparison of analyses performed by the EDATA program (1), with the best available values (2). Run time standards are underlined. Oxygen is obtained by difference.





## 6.0 AUTOMATED CONTOUR MAPPING TECHNIQUES

### 6.1 Conventional Contour Mapping Techniques

Contour mapping is an attractive approach for the display of the chemical variation in the analyzed till samples, or the elevation of the contacts between stratigraphic units. Available computer techniques were considered unsatisfactory for a variety of reasons. Computer programs using traditional surface estimation methods to produce contour maps follow a fairly standard procedure. A suitable rectangular grid is superimposed over the map area, and an elevation estimate made at each grid node. Estimation techniques vary widely, but usually consist of selecting  $n$  neighbouring data points, evaluating an inverse distance weighting function for each (giving less weight to more distant points), and then calculating a weighted average for the values obtained from these data points (e.g. Sampson, 1975). Once the grid node values are obtained, straight line (or, occasionally, curved) contour segments are obtained for a desired contour value by interpolation from the values at the corners of each rectangle. Reviews of the multitude of contouring methods are given by Rhind (1975), Walters (1969) and Crain (1970). Some of the better known publications on the subject are Shepard (1968), Cottafava and LeMoli (1969) and Dayhoff (1963).

Several problems remain with most gridding methods. One is the use of an assumed arbitrary distribution function



such as  $1/d^2$ , where  $d$  is the distance of the data point from the grid point whose value is to be determined. An excellent discussion of distance weighting functions is given in McLain (1974). Another difficulty occurs in the definition and collection of the "neighbouring" data points. This does not necessarily mean "closest" since it is desirable that the points surround the location under examination. Collection and testing of the data points is frequently expensive since there is no direct method of locating the nearest points to any arbitrary map location, and so some form of search of the whole data set is necessary. If the weighting of a data point has not reached zero when it is rejected from the set of neighbours being used to estimate elevations in a region of the map, then discontinuities in the surface may occur with weighted average techniques. Further problems are (1) the grid size defines the final resolution irrespective of data point distribution since further processing consists of interpolation between grid nodes, not data points, (2) interpolation of contour segments in a rectangle is not always unique or trivial due to the difficulty in distinguishing a north-east trending ridge from a north-west trending valley (or vice-versa), (3) fairly extensive programming may be required in order to string all the contour segments together for plotting and (4) elevations of the defined surface may not pass precisely through the data points. For topographic maps this last property is



undesirable, but for data with an appreciable error component some smoothing may be desirable. Gold (1977) discusses some aspects of the selection of a suitable contouring method.

## 6.2 Triangular Element Data Structures and Coordinates

Contour mapping using triangular element data structures possesses several advantages in dealing with the above problems. The general approach is to break the map region into triangles with data points at the vertices, obtain the slopes at each vertex and bend triangular plates to conform to these. Each of these irregular triangles is represented by an element in the data structure, with pointers to the three adjacent triangles and the three associated data points (Fig. 15). A data structure of this type that was defined manually rather than automatically is described by Heap (1972).

In order to understand the benefits of the suggested data structure it is important to understand its construction and properties. The fundamental concept is the availability of a local homogeneous coordinate system  $r_i$  ( $i = 1$  to 3) associated with each triangle (Fig. 16a). Thus an arbitrary point  $p$  at location  $(x_p, y_p)$  may be located with reference to a particular triangle with vertices at the points  $i = 1, 2, 3$  (numbered anticlockwise) by

$$[6.1] \quad r_i = d_i / D \text{ where}$$





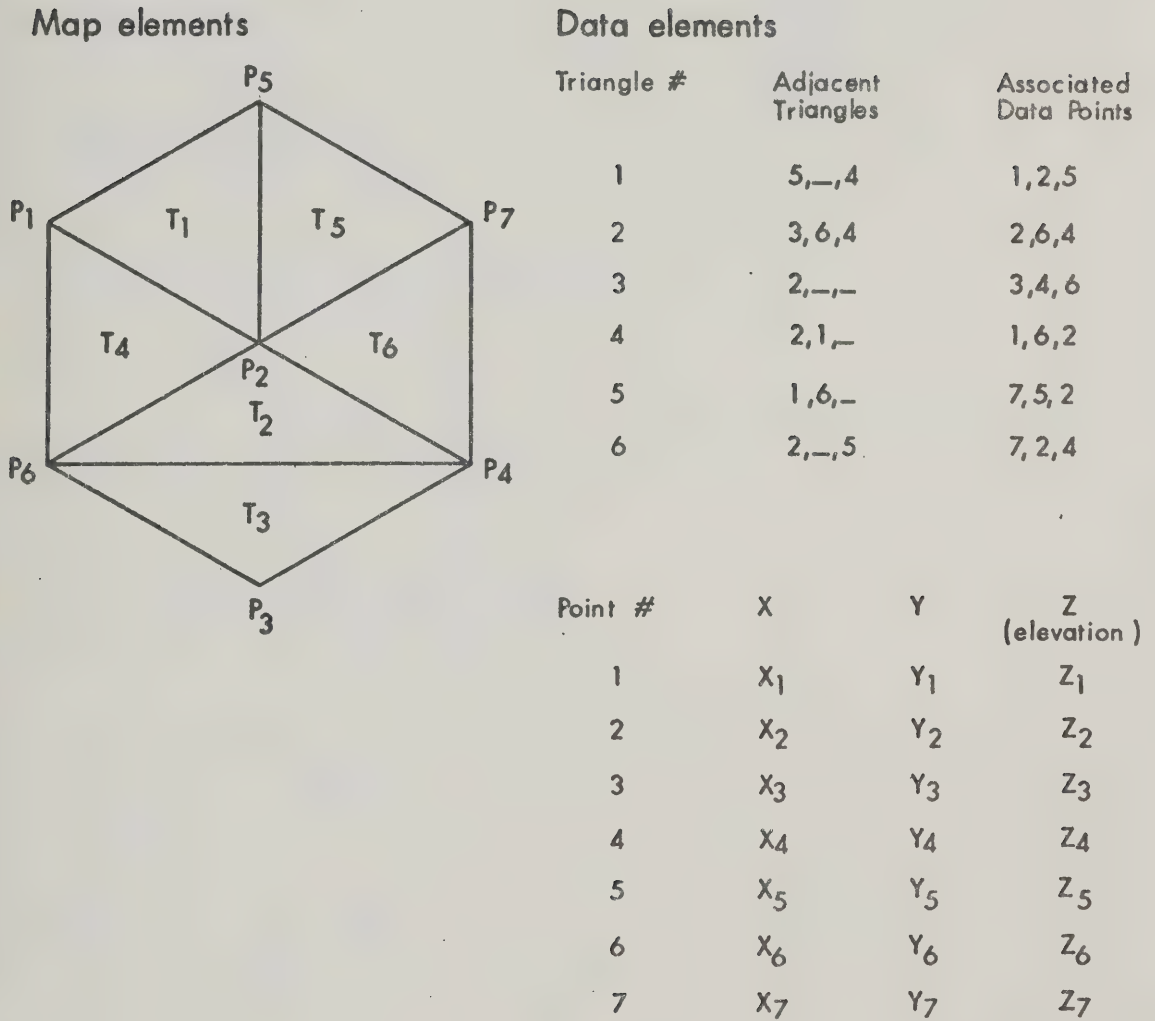
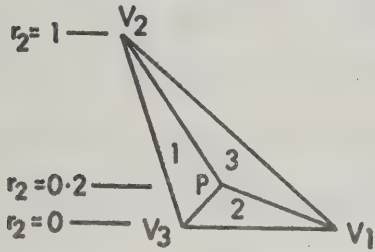


Fig.15 (From Gold et al., 1977) Triangular element data structures.

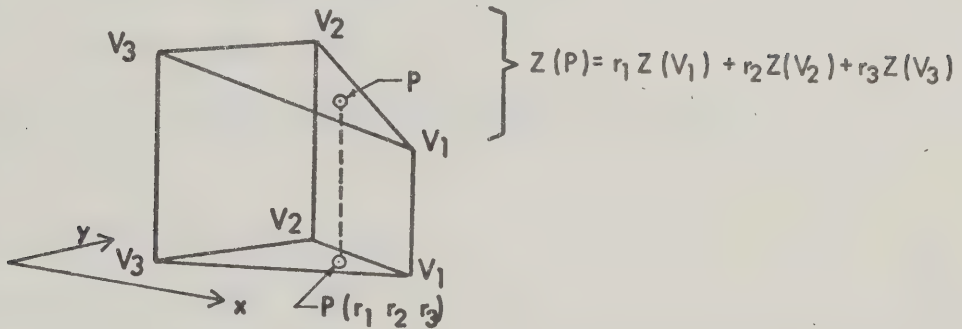


## a) Local coordinates

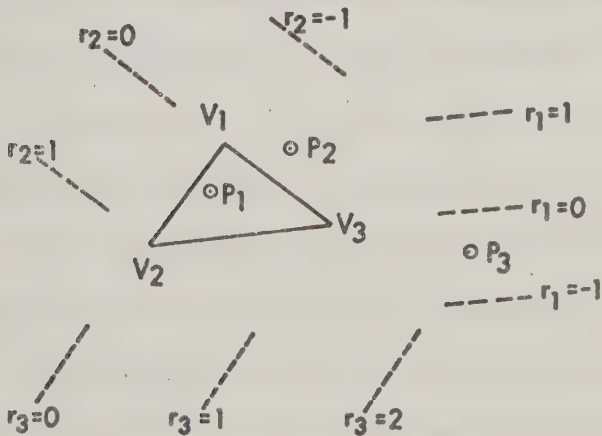


$$\begin{aligned} r_1 &= 0.4 \\ r_2 &= 0.2 \\ r_3 &= 0.4 \end{aligned}$$

## b) Planar interpolation of elevations



## c) Local coordinates and search strategies



Point	$r_1$	$r_2$	$r_3$
P <sub>1</sub>	0.4	0.4	0.2
P <sub>2</sub>	0.8	-0.3	0.5
P <sub>3</sub>	-0.4	-0.6	2.0

Fig.16 (From Gold et al., 1977) Derivation and use of a local homogeneous coordinate system for a triangle.



$$[6.2] \quad d_1 = \begin{vmatrix} x_p & y_p & 1 \\ x_2 & y_2 & 1 \\ x_3 & y_3 & 1 \end{vmatrix}; \quad d_2 = \begin{vmatrix} x_1 & y_1 & 1 \\ x_p & y_p & 1 \\ x_3 & y_3 & 1 \end{vmatrix}; \quad d_3 = \begin{vmatrix} x_1 & y_1 & 1 \\ x_2 & y_2 & 1 \\ x_p & y_p & 1 \end{vmatrix}; \quad D = \begin{vmatrix} x_1 & y_1 & 1 \\ x_2 & y_2 & 1 \\ x_3 & y_3 & 1 \end{vmatrix}$$

The value  $d_i$  may be thought of as twice the area of sub-triangle  $i$ , and  $D$  as twice the area of the reference triangle. This is the solution of the system of linear equations:

$$[6.3] \quad \begin{aligned} x_p &= r_1 x_1 + r_2 x_2 + r_3 x_3 \\ y_p &= r_1 y_1 + r_2 y_2 + r_3 y_3 \\ 1 &= r_1 + r_2 + r_3 \end{aligned}$$

In addition, if planar interpolation of elevations is required,

$$[6.4] \quad z_p = r_1 z_1 + r_2 z_2 + r_3 z_3$$

as illustrated in Fig. 16b.

This coordinate system was first devised by Feuerbach (1827) for tetrahedra and variations on it were extensively used by geometers of that century. Each of these three coordinates ( $r_i$ ) has the value 0 when  $p$  falls on the base line  $i$ , and 1 when  $p$  falls on a line parallel to base line  $i$ , but passing through the opposite vertex  $i$ . The three coordinates sum to unity, hence any two define the third. All three coordinates are positive if  $p$  falls within the triangle, and if  $p$  falls outside, the signs of the coordinates indicate the direction to go looking for a circumscribing triangle.



### 6.3 Data Structure Generation and Optimization

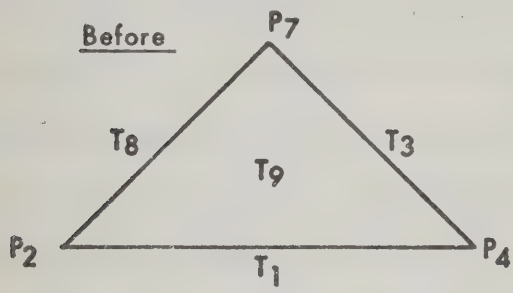
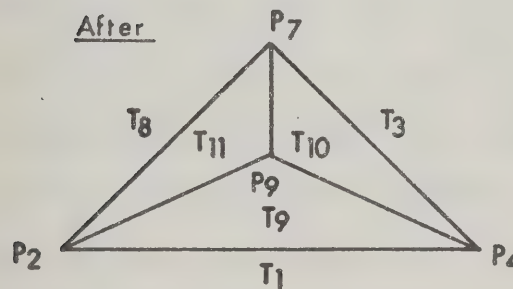
It is clear from Fig. 16c that, if a data structure relating adjacent triangles exists, a search algorithm is easily devised to start at any arbitrary triangle and search through a set of adjacent triangles until the one enclosing the unknown point is located. This in itself suggests a method for generating such a data structure. A "universal" triangle, large enough to enclose the map area, is first generated. The data points are inserted one at a time and the enclosing triangle for each is located. This triangle is then divided into three new triangles with the new point at a vertex common to all three, and all pointers are then updated (Fig. 17a).

The resulting mesh of triangles is continuous over the map area but the triangles produced may have undesirable shapes. An optimization routine is then used to test each triangle against each of its neighbours. The quadrilateral thus formed, if convex, may be divided in two ways, and it is the purpose of the optimization criterion to choose one of them. If the current division is the poorer of the two then the quadrilateral is divided in the alternative manner and the data structure pointers updated as in Fig. 17b. In this manner every triangle pair in turn is examined to see if it is acceptable and, if not, the alternative partition is taken. This procedure is repeated until an entire pass through the data structure produces no changes. By the judicious setting of flags the optimization of even a few





## a) Inserting new data points

Map elements		Data elements		
Before		Triangle #	Adjacent Triangles	Associated Data Points
		9	1, 3, 8	7, 2, 4
<p>After</p> 		9	1, 10, 11	9, 2, 4
		10	9, 3, 11	7, 9, 4
		11	9, 10, 8	7, 2, 11

## b) Quadrilateral optimization

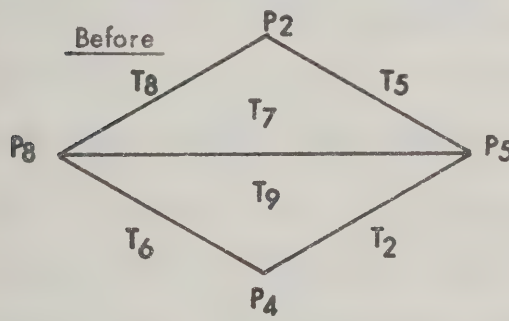
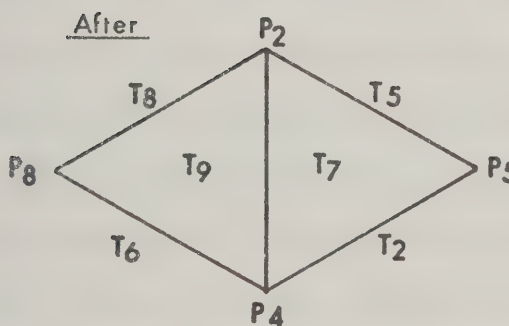
Before				
		7	9, 5, 8	2, 8, 5
		9	7, 6, 2	4, 5, 8
<p>After</p> 		7	2, 5, 9	2, 4, 5
		9	8, 6, 7	4, 2, 8

Fig.17 (From Gold et al., 1977) Triangular data structure generation and optimization.



thousand triangles remains an economic proposition, the insertion and optimization of 1129 data points requiring 0.22 and 1.92 seconds respectively on an Amdahl 470/V6 computer for a test data set. These flags are set to define triangle pairs that have been tested and found satisfactory, and hence need not be tested again until one of the two triangles has been redefined. Fig. 18 (solid line) illustrates the rate of convergence of the iterative optimization. The natural logarithm of the number of switches made decreases by close to one third on each iteration, and hence the number of iterations approximates  $3 \ln(T)$  where  $T$  is the number of triangles. This is a worst-case condition since the dashed line shows the convergence rate for a second batch of data points inserted after the first set had been optimized. This approach seems to have considerable economic advantages over the methods of Frenkel and Gill (1975) and McLain (1976) in which the mesh is built up agglomeratively from some nucleus by testing all neighbouring points to find the one most suitable to be added at a specific mesh location. Lawson (1977) generates a mesh by first finding the convex hull of the set of data points and agglomeratively building the mesh within this. Optimal triangulation (as defined in Section 6.4) is achieved by testing triangle pairs, as is done in this work. The use of a fast triangulation algorithm, as suggested here, may permit the contouring of moderate to large data sets several times faster than by gridding methods.



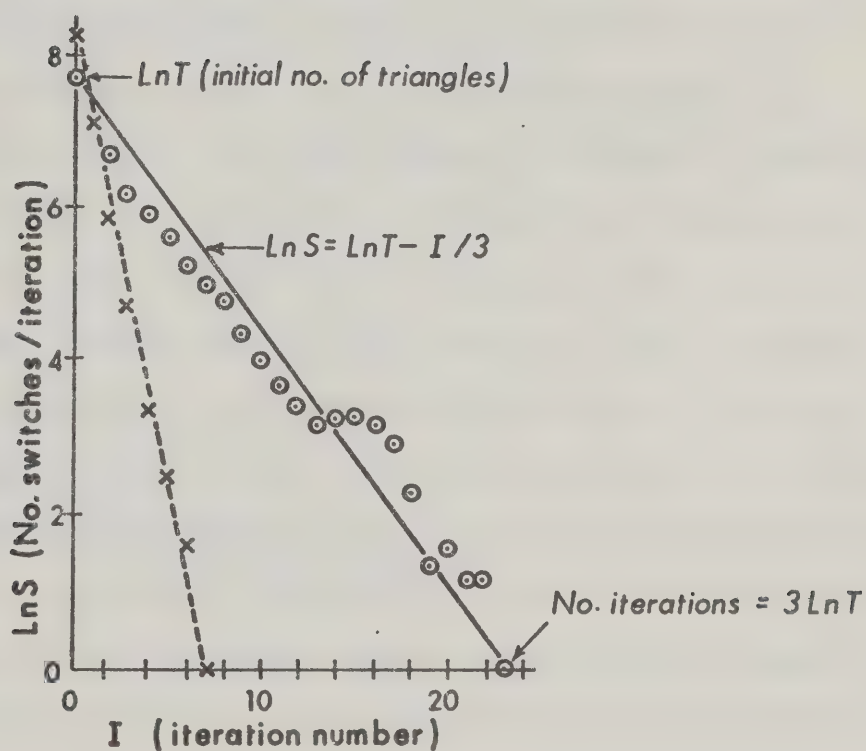


Fig.18 (From Gold et al., 1977) Rate of convergence of optimization of a triangular mesh.



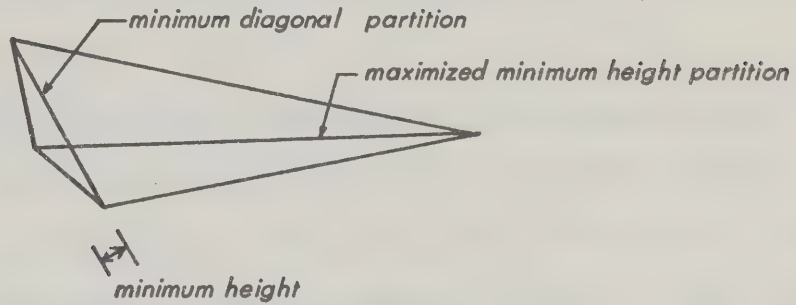


#### 6.4 Choice of an Optimization Criterion

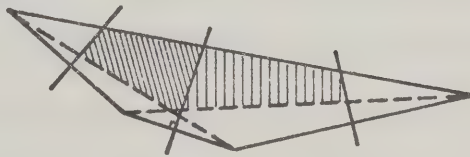
An obvious question remains - what is our optimization criterion? Rhind (1975, p. 297), in reviewing spatial interpolation techniques, states his ignorance of any "...fundamentally sound, cheap and easily-applied partitioning theory". He does however, mention an Israeli firm's approach of a single pass through the lattice to minimize the total length of the triangle sides, as well as a commercial package (SCA, 1970) which use the "...best overall approximation to equilateral triangles" (Rhind, 1975, p. 297). Gold (1976a) similarly optimizes by choosing the partition of the previously mentioned quadrilateral with the shortest diagonal. This however proves to be somewhat unsatisfactory as triangles with an unnecessarily small height could sometimes be produced. A more satisfactory criterion turned out to be the maximization of the minimum triangle height. This height is measured perpendicular to the diagonal bisecting the quadrilateral. Thus two heights are obtained for a given partition of the quadrilateral, and the smallest of these is retained. This height is then compared with the equivalent figure for the alternative partition of the quadrilateral, and the final partition chosen is the one having the largest value for this minimum height. Fig. 19a illustrates the difference between this and the shortest diagonal criterion. The minimum diagonal criterion produces the left-hand triangle: however, the height of this triangle measured perpendicular to the



a) Differences between optimization criteria



b) Imperfect optimization



c) Equivalence of criteria

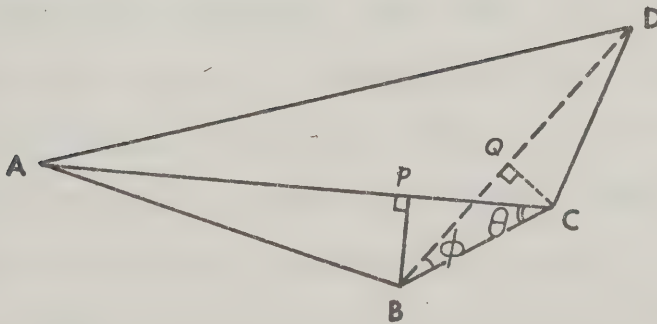


Fig.19 (From Gold et al., 1977) Optimization criteria. (a) shows the difference between two proposed criteria, (b) indicates areas for which triangulation is not optimum and (c) shows the geometric relationships between two proposed criteria.



relevant diagonal is the smallest of the four possible triangles, and hence this partition is rejected when the maximized minimum triangle height criterion is used. Lawson (1977) maximizes the minimum angle, instead of the height as is done in this work.

McLain (1976, p. 179), referring to Pitteway (1973) states that "An optimal partition is one in which, for any point within any triangle, that point lies at least as close to one of the vertices of that triangle as to any other data point." He then gives an algorithm to choose from among the candidate points. While the definition is excellent it cannot always be achieved, as shown in Fig. 19b. Whichever way the quadrilateral is partitioned, one of the shaded zones will not be "optimal" since any point within that zone will be closer to a data point outside the triangle in which it occurs than it will be to any data point forming a vertex of the containing triangle. Thus any point in a shaded zone is closer to a "bad", or undesirable, data point than it is to any "good" data point. The worst possible situation occurs at points P or Q in Fig. 19c, at the closest point in the shaded zone to the "bad" data point.

With this situation in mind Pitteway's optimal partition may be rephrased as an optimization criterion. This is defined as minimizing the ratio of the distance to the closest "good" point over the distance to the closest "bad" point, measured from the worst-case position in the triangle. If the closest "good" points for P and Q are C and





B respectively then the relevant ratios are  $PC/PB$  and  $QB/QC$ , or  $\cot \theta$  and  $\cot \phi$  respectively. Since  $\sin \theta = PB/BC$  and  $\sin \phi = QC/BC$  it is clear that maximizing the minimum height (i.e. choosing the larger of  $PB$  or  $QC$ ) fulfills the theoretical optimization criterion. If the nearest "good" point is not on the suggested diagonal then  $PC$  and  $QB$  form upper limits to  $PD$  and  $QA$ , again fulfilling the required optimization criterion (Gold, 1977).

It is, of course, sometimes advantageous to modify or build the triangulation by hand. This provides the facility to modify the surface on the basis of additional information, such as continuity of valleys and ridges, which is normally difficult to build into the mapping program. A good description of this problem under the heading of "surface specific points" is contained in Peucker (1972).

### 6.5 Mapping with Triangular Elements

Once the data structure has been generated, its benefits in contour mapping, for example, become apparent. The problem of identifying the neighbouring data points for any arbitrary location is readily solved. Its three immediate neighbours form the vertices of the circumscribing triangle and, if desired, the three additional vertices from the three adjacent triangles may be included. This approach forms a very happy combination of the requirements for neighbours to be both close to and surrounding the point under consideration. If a simple contouring method is





sufficient, each triangle may be considered to generate a function consisting of a plane passing through the three data points, as in Bengtsson and Nordbeck (1964). The surface made up of the set of triangular planes is readily contoured, there is no difficulty defining the domain in which a given set of data points define the surface, and the resolution of the final map is a function of the local data point density. The surface is, of course, discontinuous in the first derivative across triangle boundaries and its contour lines are therefore angular. For many applications, especially where the data point distribution is very dense and/or the final product required is the elevations at the nodes of a grid, this is an extremely efficient technique.

Alternatively, conventional contouring techniques could be used to produce a grid of values, merely using the triangular element data structure for collection of neighbouring data points. This would, however, neglect several advantages of the triangle approach. Therefore, in this work a method has been devised to interpolate within a triangle using the local coordinate system of Fig. 16a. In this approach, the first step is to estimate a best-fit plane, (constrained to pass through the data point), at each datum location. The surface is then estimated for each triangular domain in turn using only the three planes at the vertices. These three planes may be expressed in a form compatible with the triangular coordinate system as  $E_{ij}$ , the elevations of each of the three planes  $i$  at each of the



three vertices  $j$  (Gold et al., 1977). If field measurement of the slope at data points is available (for example with dipmeter readings from drill holes) this may be used instead of the estimated best-fit plane. Use of slope information greatly increases the information content of the surface. It is suggested in McLain (1974) that the use of  $x$  and  $y$  slopes as well as the elevation information at  $n$  data points is approximately equivalent to  $2.5n$  data points with elevations alone. This approach was also used in the SURFACE II contouring package (Sampson, 1975), although in both cases interpolation was made onto a square grid.

#### 6.6 Requirements of an Interpolating Function

Within each triangle an equation can be developed using the local coordinate system and the nine values  $E_{ij}$  for a surface possessing certain specified properties. These properties are similar to the surface properties for any contouring method. The first is that the surface passes through the three data points - a condition not met by many moving average and gridding methods. The surface should also possess the measured or estimated slope at each data point. This condition is rarely met in conventional contouring programs, and with moving average techniques the slope at data points may be zero. It follows from this that extrema should be free to occur between data locations. This is the case with global polynomial methods, although the values may be unreasonable. The SURFACE II program, (Sampson, 1975)



achieves flexibility of extrema locations by weighting the estimate produced by a best-fit plane at each data point. A fourth requirement is that the surface is continuous between domains, or regions of the map in which a specific function is used to interpolate the surface. For triangulation techniques this domain is obviously defined as the triangle. For conventional methods the definition is less obvious, but in fact it is the region of the map where a particular set of "neighbouring" data points are used to estimate the surface. Failure to give a point a zero weight before it leaves this set results in unexpected breaks in the surface and kinks in the contour lines. A last requirement is that the slope of the surface is continuous between domains, providing smooth contour lines across domain boundaries.

### 6.7 Shape Functions

Several attempts have been made to meet all these criteria, especially those of continuity of slope. Suggestions for obtaining an interpolated elevation estimate by weighting each of the three planes at the vertices of the triangle by the square of its associated local coordinate ( $r$ -value) (Gold, 1976b) or weighting a quadratic curve at each vertex by  $r^3$  (McLain, 1976) have been made. Both claimed that the slope of the interpolated surface did not change abruptly when triangle boundaries were crossed, and in both cases this appears to be incorrect for reasons given by Zienkiewicz (1971, p.175). Differentiation of either





Gold's or McLain's equations in Cartesian coordinates, although tedious, clearly shows the misconception. For example, the function slope at edge 1 of a triangle is indeed independent of the coefficients of the plane or quadratic surface associated with the opposing vertex 1. However, the function slope is not independent of the location of this vertex since the weighting functions used on the remaining two surfaces associated with vertices 2 and 3 are dependent on the local coordinates  $r_2$  and  $r_3$  which have been shown previously (equations [6.1] and [6.2]) to involve the values  $x_1$  and  $y_1$ . If moving vertex 1 changes the slope at and perpendicular to side 1 then clearly slope continuity cannot be expected when passing into the adjacent triangle. The effect is particularly marked when angle 2 or angle 3 of the triangle is obtuse. If a function (in this case surface slope) is to be continuous across triangle boundaries it must have no dependence on any property that differs between the two triangles. Thus slope continuity can be achieved only when all slope directions at the boundary are free of all properties of the opposite vertex - vertex 1 in this example. This cannot be achieved by any simple weighting of vertices by local coordinates since the weighting at the boundary depends on the location of the opposing vertex. This is briefly discussed in the Errata to McLain (1976).

A more helpful approach is to look at the work done on finite element analysis in engineering. Zienkiewicz (1971)



and Irons (1969) are particularly useful, the first of these authors giving a relatively straightforward "shape function" which is used here. This function can be expressed as

$$[6.5] \quad z_p = \sum_{i=1}^3 \sum_{j=1}^3 w_{ji} E_{ij} ,$$

although not used in this form by Zienkiewicz. Here  $z_p$  is the elevation of the surface at an arbitrary location  $p$  and the  $E_{ij}$  are the coefficients of the planes associated with the vertices of the triangle, as defined previously. If  $i \neq j \neq k \neq i$  then

$$[6.6] \quad w_{ji} = r_i^2 r_j + 0.5 r_i r_j r_k, \text{ and } w_{ii} = r_i - w_{ij} - w_{ik}$$

where  $i, j$  and  $k$  refer to the three local coordinates.  $r_k$  is used here instead of  $1 - r_i - r_j$  in order to preserve the symmetry of the operations performed using homogeneous coordinates.

This function possesses the correct slope at data points but does not necessarily provide continuity of slope across triangle boundaries. Zienkiewicz, however, suggests an empirical remedy. If an assumption is made about the slope normal to a side at the mid-point of that side, then the function may be corrected to conform to this. Thus  $S_i$  may be defined as the difference between the function slope (perpendicular to the boundary) and the average of the slopes of the two planes associated with that side, and  $v_i$  as a function having zero value or slope at sides  $j$  and  $k$  and unit slope at the mid point of side  $i$ . Here

$$[6.7] \quad v_i = 4(1+r_i) r_i r_j^2 r_k^2 / (r_i + r_j) (r_i + r_k) .$$

Addition of  $\sum_{i=1}^3 v_i S_i$  to the previous function will provide



the necessary slope continuity. Other assumptions can be made about the mid-side slopes, producing a variety of adequate shape functions.

#### 6.8 Contour Mapping with the Interpolant

The surface thus produced is continuous, and is continuous in the first derivative, across all triangle boundaries. In order to plot the surface, each main triangle is divided into  $N^2$  equal sub-triangles and  $Z_p$  calculated at each node. Linear interpolation of contour segments within each sub-triangle is straightforward since no saddle-point problem exists once the basic triangulation has been defined. The resolution of the resulting map is a function of  $N$  and the local data point density. Using this approach a contour map was produced (Fig. 20) of the strength of the vertical component of the earth's magnetic field, sampled at 26 sites in Alberta (Bannister, 1977). Fig. 21a shows a plot of  $\sin(x)\sin(y)$  where slopes are known at the data points. Fig. 21b is the resulting plot when slopes must be estimated, as is usually the case. Apart from boundary effects the results agree quite well. These figures illustrate the satisfactory performance of the smooth interpolating function under real as well as mathematically defined conditions.

It is of interest to compare the results of the triangulation method with contour maps produced using the grid approach. Fig. 22 shows the bedrock surface of the





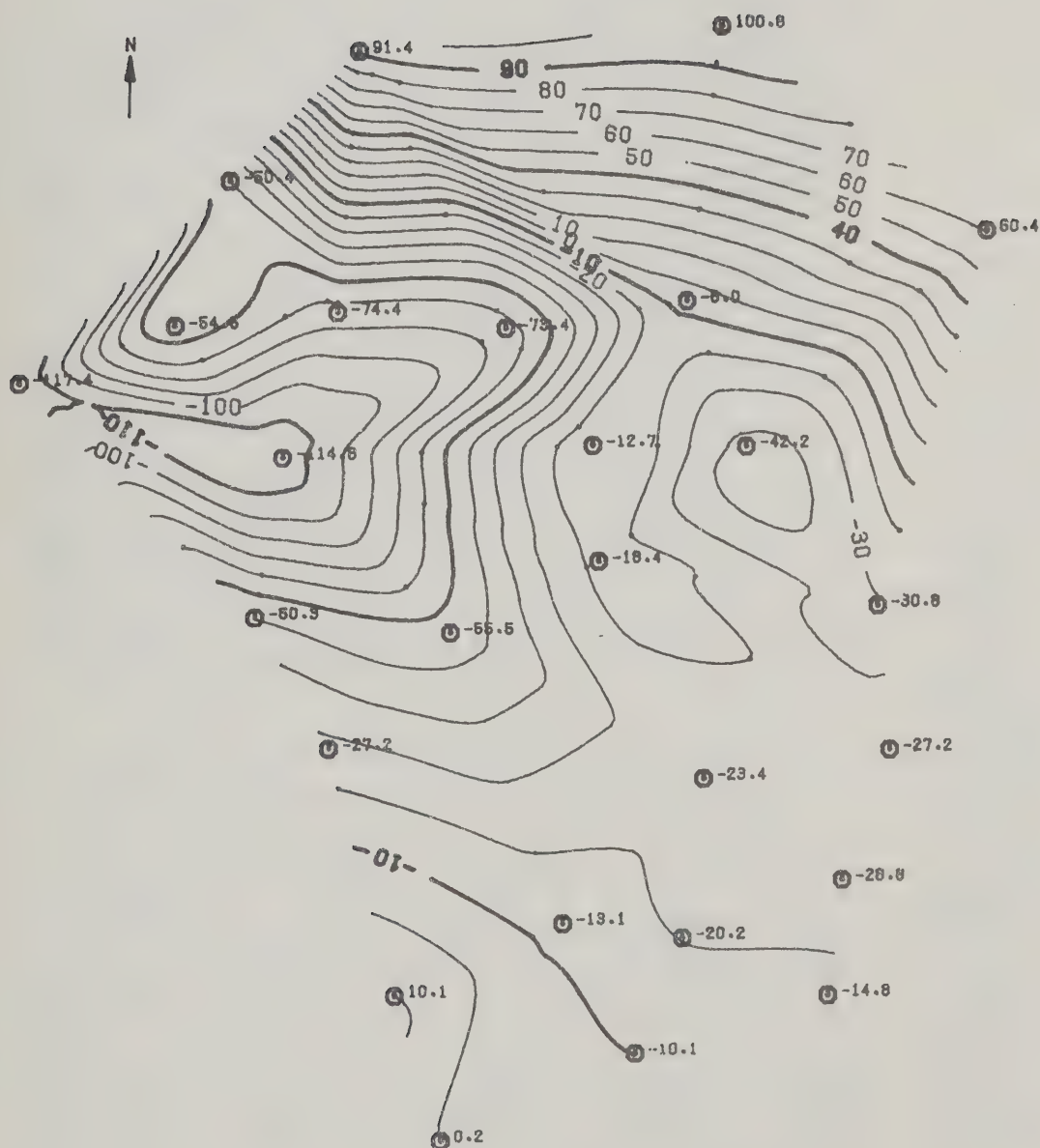
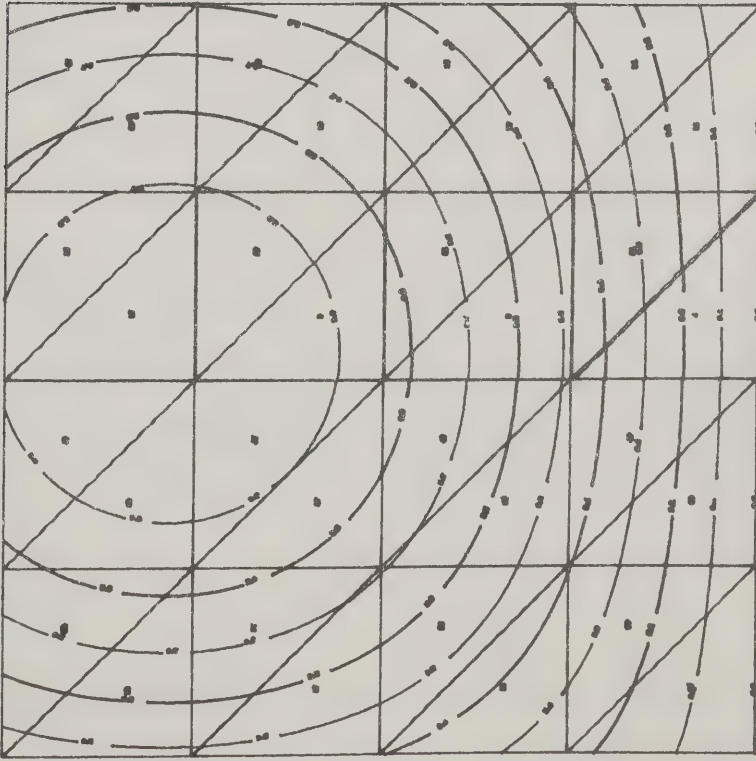


Fig. 20 (After Bannister, 1977) An example of triangulated contour mapping - a magnetic field map.

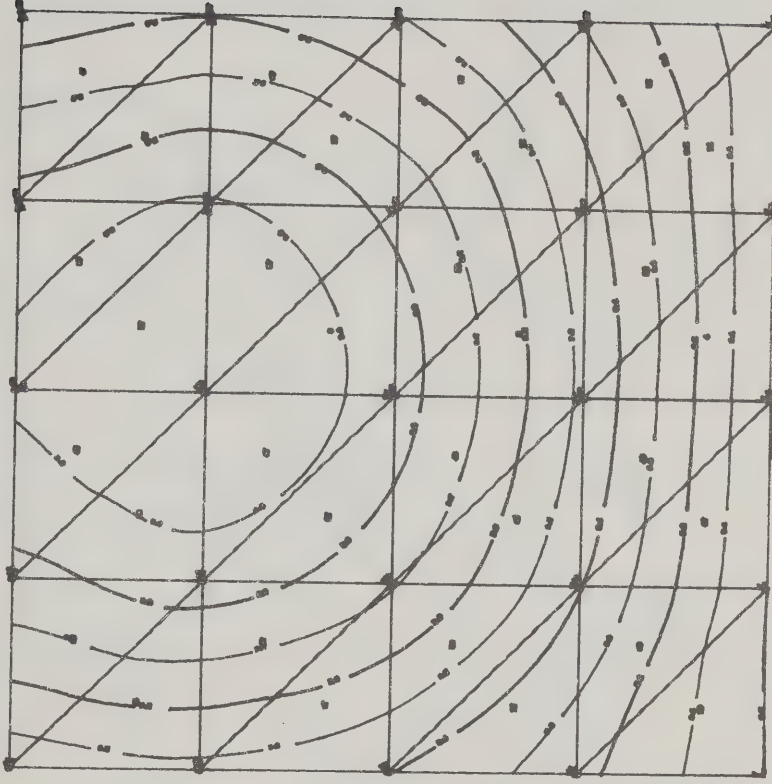




a) Known slopes at vertices



b) Estimated slopes at vertices

Fig. 21 (From Gold, 1977) Plot of  $Z = \sin(X) \sin(Y)$



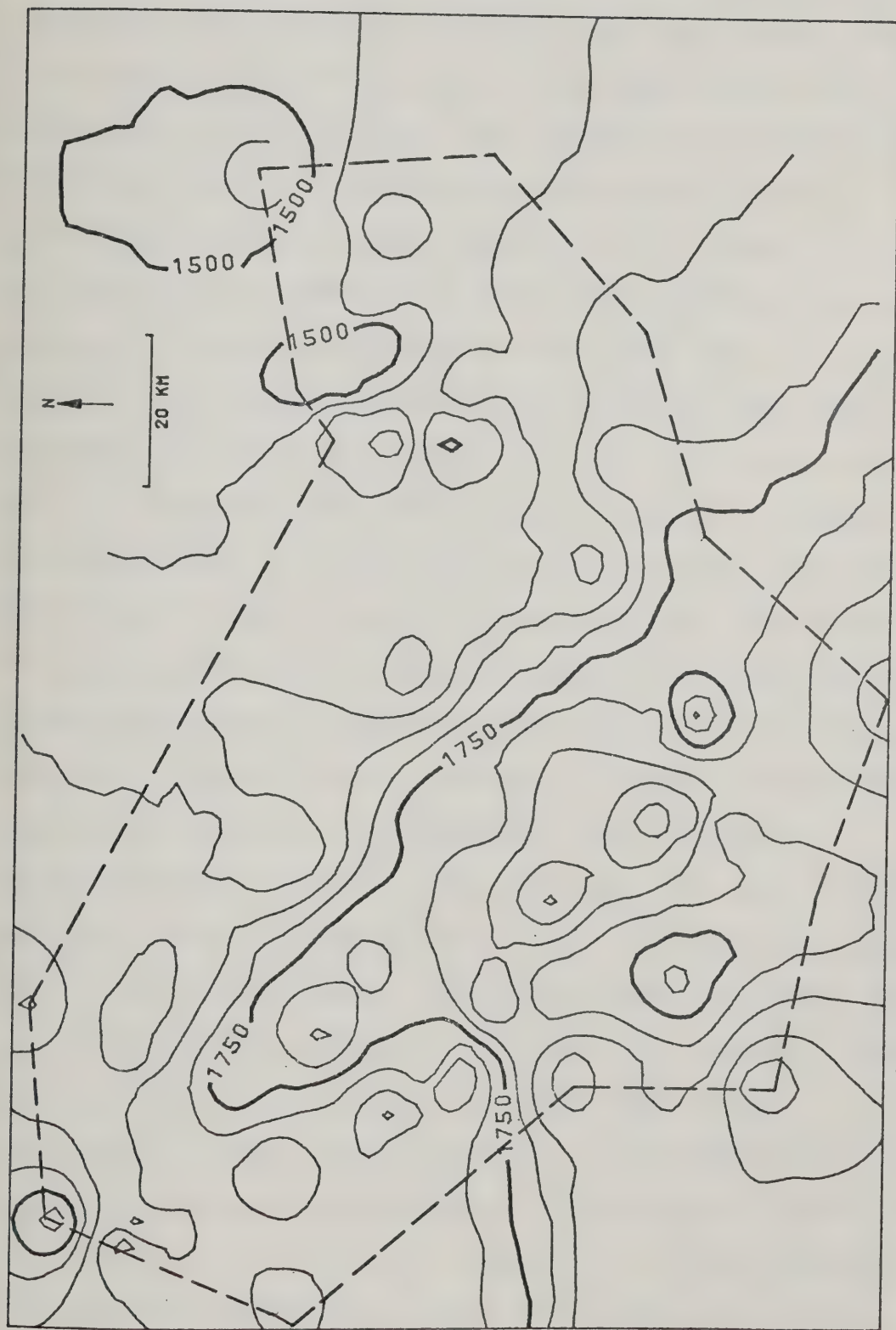


Fig. 22 Bedrock surface elevation in the map area, contoured using gridding and distance - weighted elevations at data points. (50 ft. contour interval).



study area produced by the SURFACE II contouring package, (Sampson, 1975) from the 78 drill hole data points used in this study. Seven of the drill holes used (434E, 725E, 765E, 809E, 952E, 965E and 774E) were not drilled to bedrock and hence the bedrock elevations at those locations were estimated visually from neighbouring data points. A 50 by 50 grid was used in association with  $1/d^2$  weighting of a fixed number of nearest neighbours to each grid node. Fig. 23 was produced in a similar fashion, but weighting the elevation estimate derived from the best fit plane at each data point. Fig. 24 shows the same data contoured using the method developed here. If these maps are compared with the hand drawn map of Fig. 25 which is based on human interpretation as well as additional information it will be seen that the triangulation method delineates the three major channels much better than the grid approach. Although the hand drawn map takes advantage of judgemental data not readily quantifiable for computer use, the major features are partly resolved by the triangulation of the 78 data points alone. There are, however, still several "bridges" where the data point distribution imposed a triangulation forcing a ridge to cross what is in reality a valley. This is the traditional saddle point problem in a different form, and these occurrences define regions of inadequate sampling for the features to be resolved. However, if a hand generated triangulation is defined, based on the additional human knowledge (Fig. 26), then the surface may be forced to





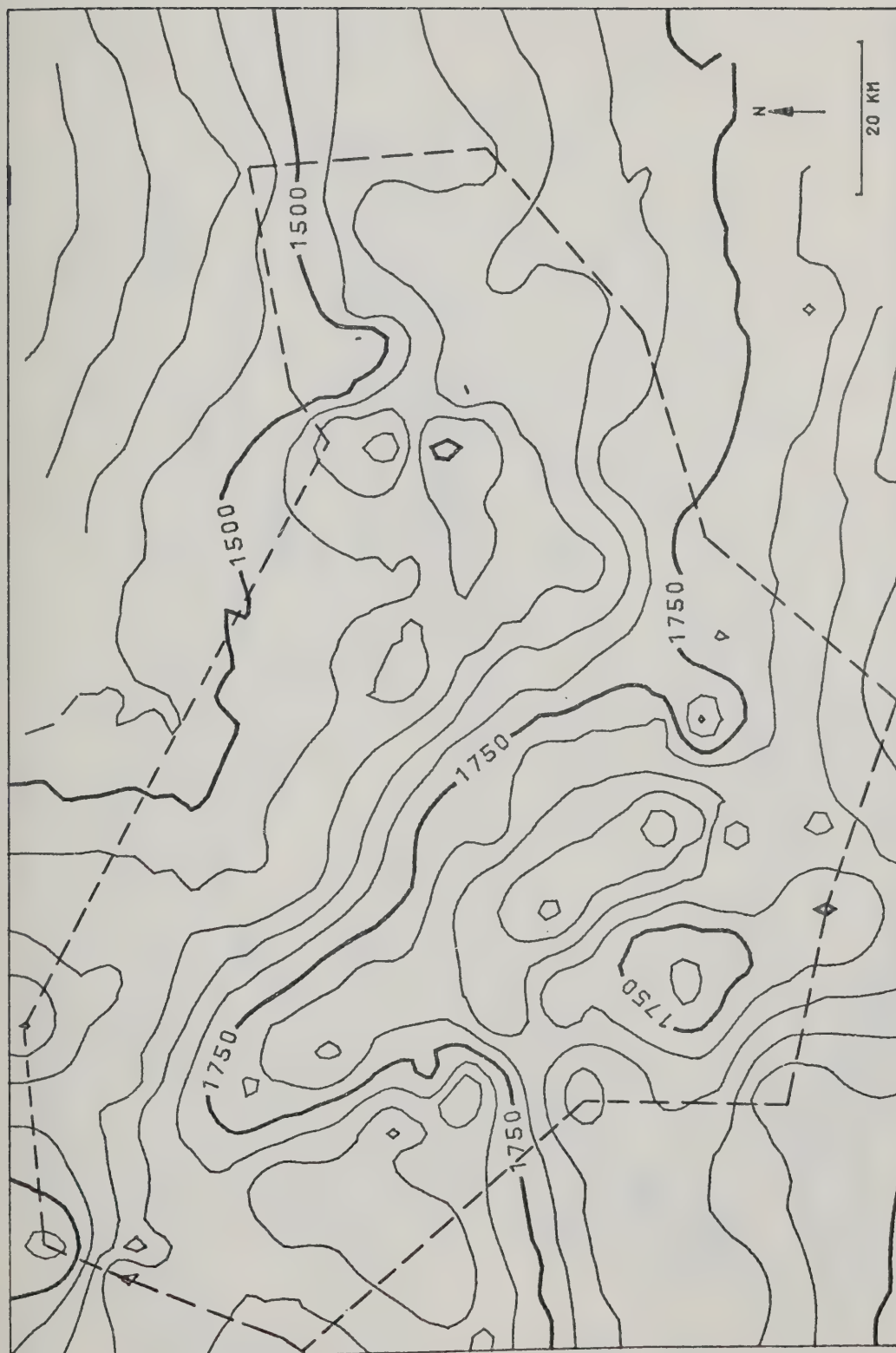


Fig. 23 Bedrock surface elevation in the map area, using gridding and distance - weighted best - fit planes at data points. (50 ft. contour interval).



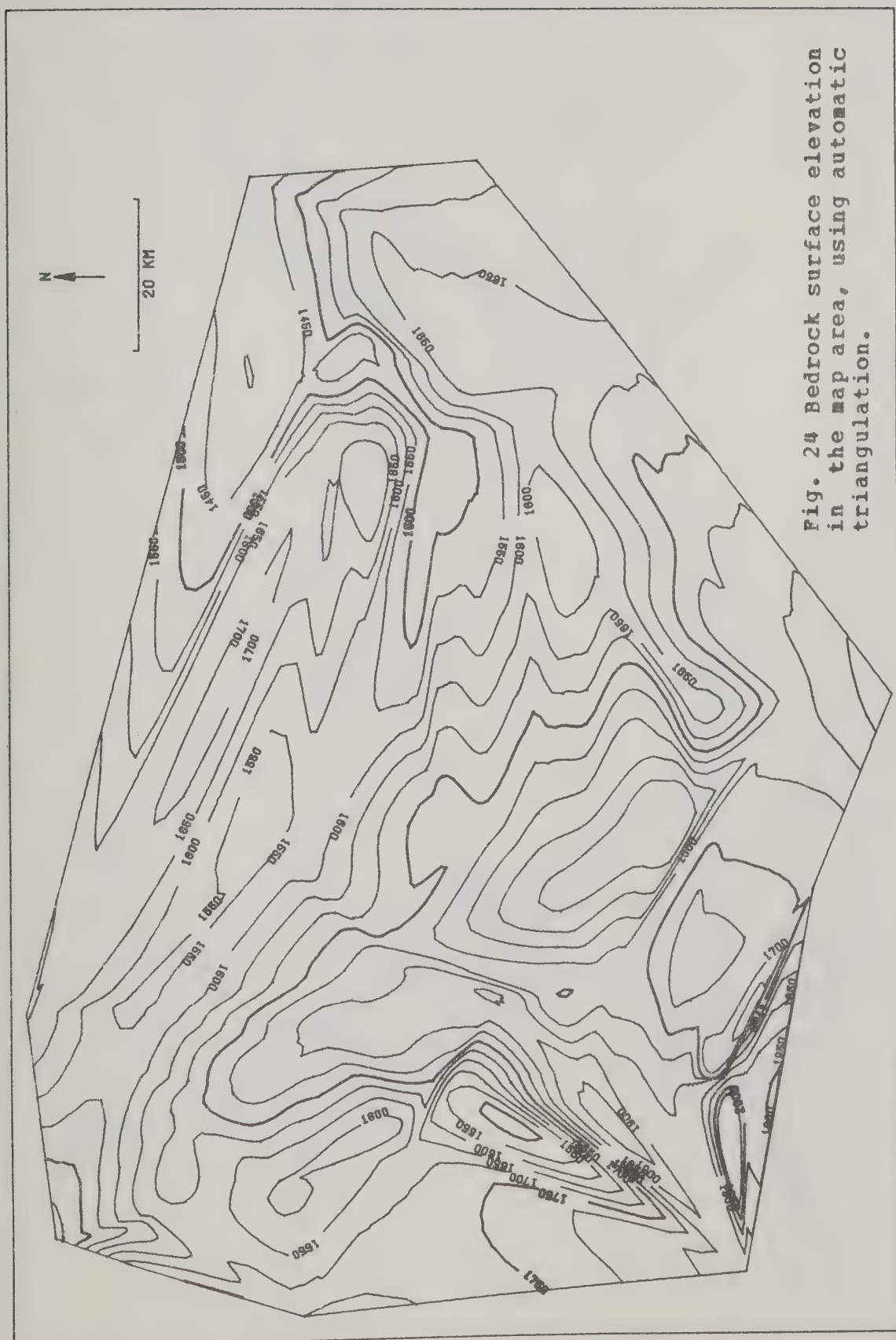


Fig. 24 Bedrock surface elevation in the map area, using automatic triangulation.



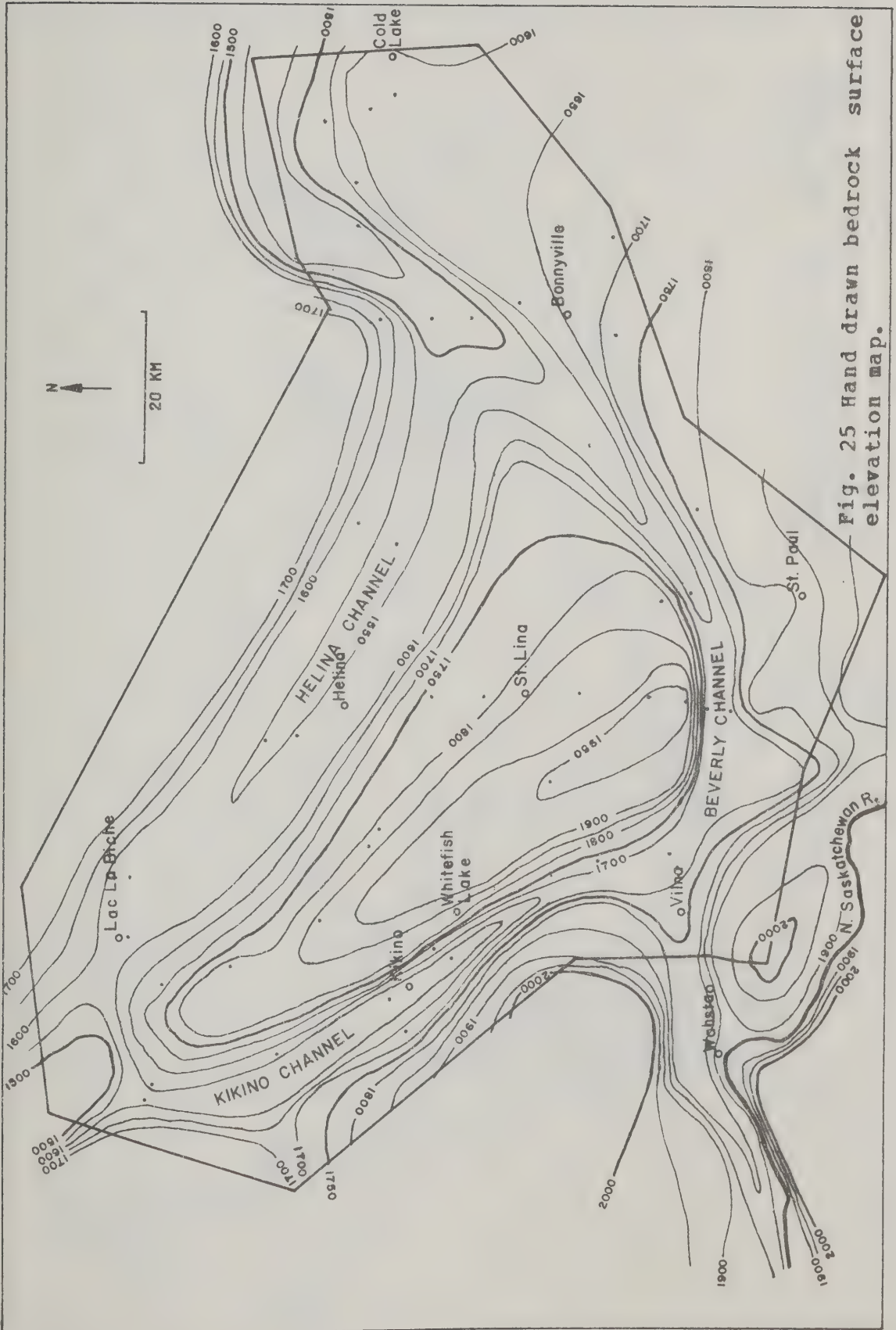


Fig. 25 Hand drawn bedrock surface elevation map.





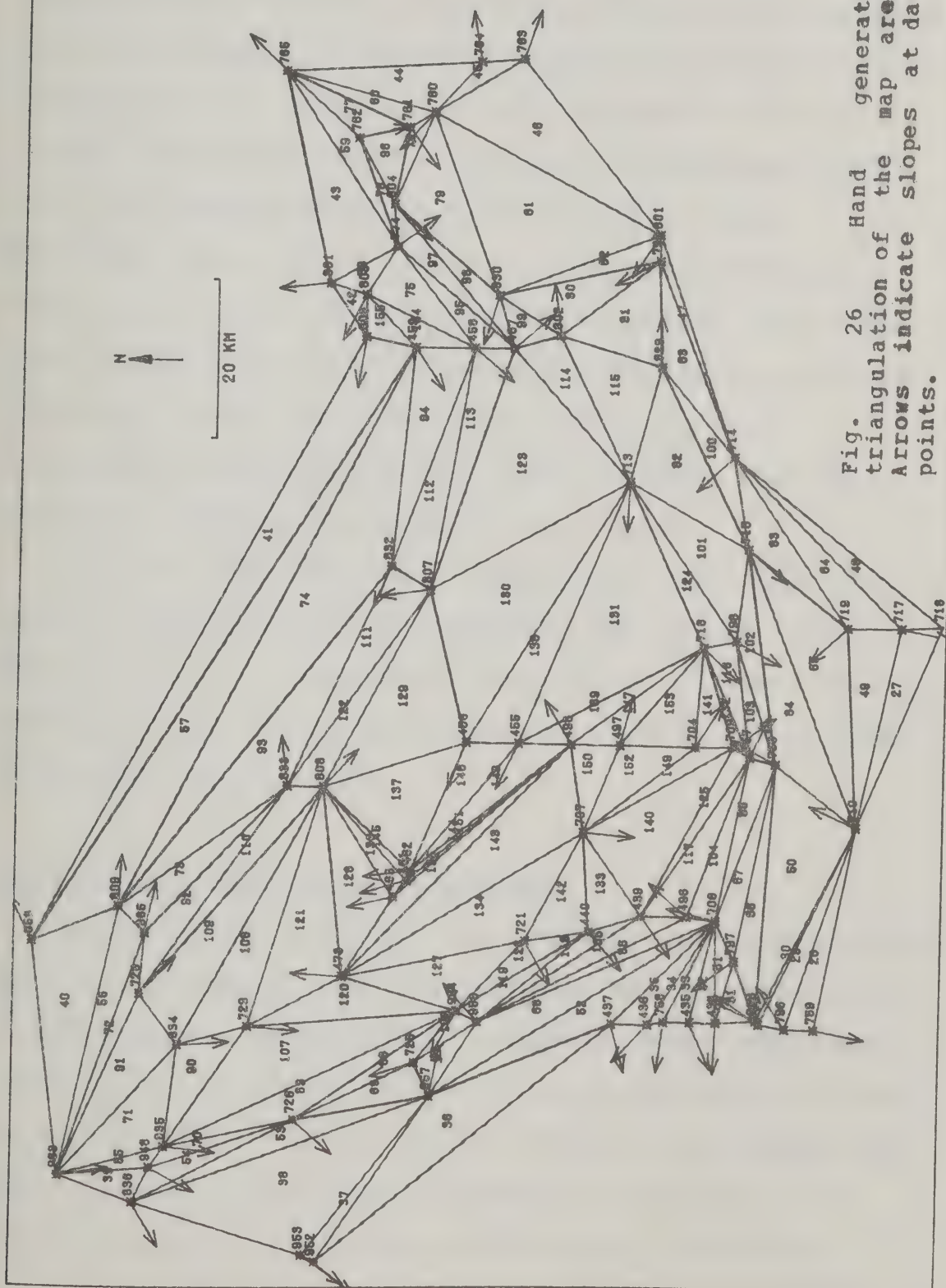


Fig. 26 Hand generated triangulation of the map area. Arrows indicate slopes at data points.





conform to the predetermined interpretation (Fig. 27).

This approach is closely related to the idea of surface specific lines expressed in Peucker (1972). Surface specific points were defined as points that furnish more information than merely their coordinates, such as peaks, pits and passes. Surface specific lines consist of ridge and valley lines, and connect surface specific points, ridges connecting peaks and valleys connecting pits. Passes are formed at the intersection of ridge and valley lines. This writer knows of no method for analytic definition of ridge and valley lines, and hence these must currently be identified by the operator and either inserted into the program or added to the map at a later stage.

At least one commercial organization (SCA, 1970) is known to use triangulation techniques for contour mapping but details of its methods are proprietary and hence remain unknown to this writer. No account is taken of the problems of surface specific lines.

#### 6.9 Extension into Higher Dimensions

McLain (1976) suggests the possibility of interpolation in higher dimensions but unfortunately the author has found no interpolating function that guarantees slope continuity between domains. Simple planar interpolation however is both useful and direct: it is necessary merely to generalize the mesh construction for higher dimensions, and a method is proposed below. Harbaugh and Merriam (1968) generated



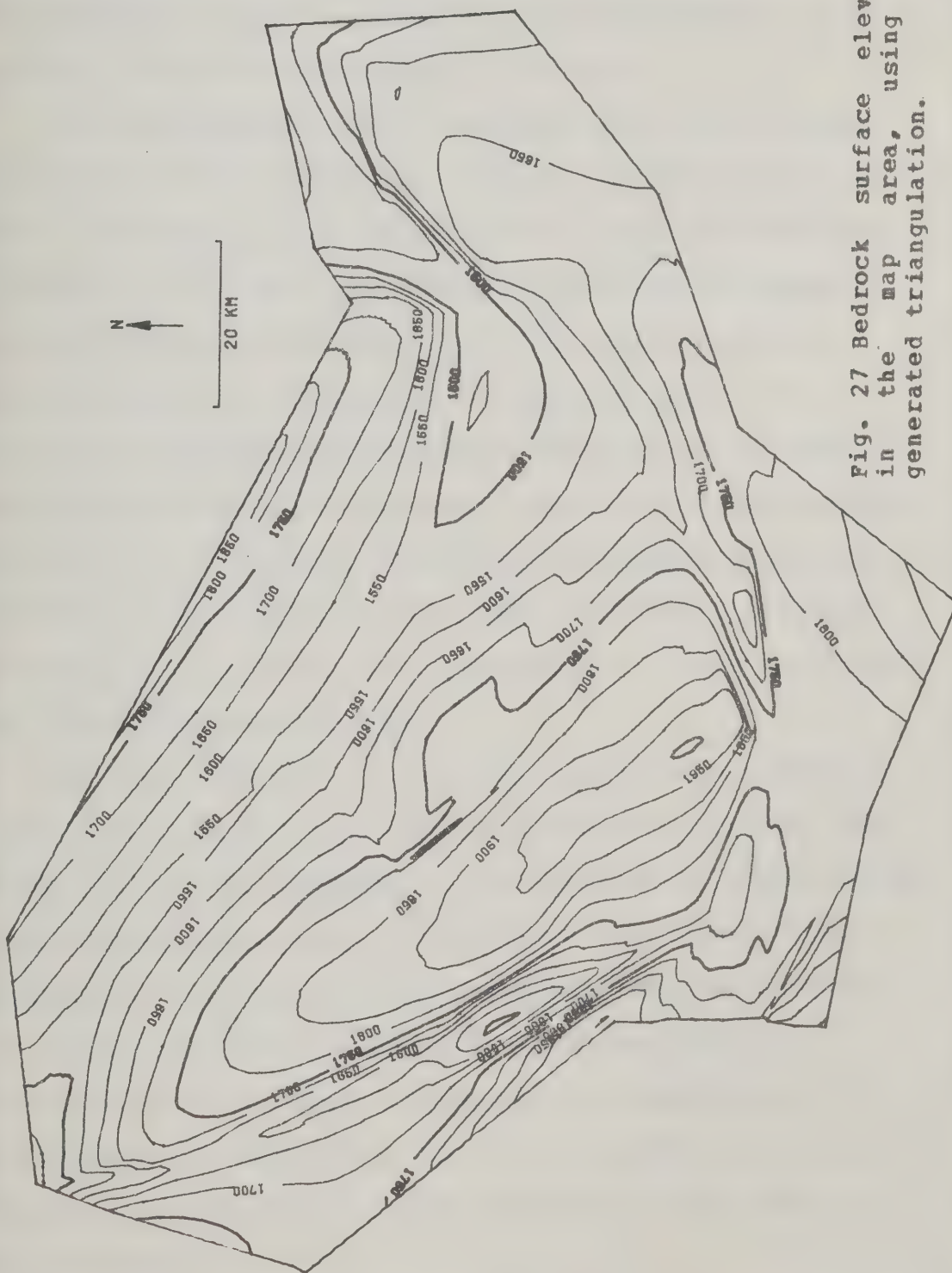


Fig. 27 Bedrock surface elevation in the map area, using hand generated triangulation.



hypersurfaces by performing trend surface analysis on crude oil gravity values from samples with locations given in three dimensions. Similar applications concerned with contouring the copper concentration in an orebody, for example, could be of geological interest.

In  $n$ -dimensional space a simplex (i.e. the simplest polyhedron of that dimension) has  $n+1$  vertices and hence  $n+1$  local coordinates, each involving  $n+1$  order determinants. Insertion of new data points into a universal simplex is therefore direct. Optimization of the data structure, however, has to be rephrased. An adjacent pair of  $n$ -dimensioned simplices will have  $n$  vertices around their common face, and two independent vertices. The alternative structure is to connect these two independent vertices, producing  $n$  simplices around this central axis like the segments of an orange. This alternative is only possible if the simplex pair is convex.

If this condition is met the optimization choice is between the simplex pair and the  $n$ -simplex "orange", and either set may be switched to the other on the basis of the maximized minimum height optimization criterion. It is interesting to note that maximizing the minimum height is one of the few possible criteria in higher than 2-dimensional space, since minimizing the diagonal is meaningless and maximizing the minimum volume is of little benefit when the number of simplices is not the same for both alternatives.





## 7.0 LABORATORY RESULTS

### 7.1 Differentiation of Till Units

Appendix 4 contains diagrams of each drill hole in the map area. In the centre of each diagram is a reduced copy of the electric log for that hole, with broad descriptive categories of the material, as well as the location of oxidized zones, in the left part of the central column. The calcium carbonate equivalent and bulk chemistry of the silt+clay fraction are expressed as histograms to the right. The far left shows the grain lithology results, and the near-left the grain size figures. These last are displayed by plotting (from right to left) a dark line for the 10 to 20 mesh fraction, a gap for the 20 to 40, a line for the 40 to 60 fraction, another gap for the 60 to 140, another line for the 140 to 230 mesh size fraction, a gap for the silt and a line for the clay fractions. In this fashion much of the information needed to distinguish the various till units is assembled in one place. The most important additional property, lateral continuity, was incorporated by extensive correlation between drill-holes, using the properties displayed in Appendix 4 as aids in matching till units.

Examination of the plots in Appendix 4 indicates the presence of at least three tills. An upper till (till 1) of moderate ( 3-6%) carbonate content covering the whole map area is underlain in the eastern part of the map area by a till of higher ( 5-10%) carbonate content (till 2). Although



readily identifiable locally as moderate or high carbonate tills, these terms vary in meaning across the map area, all tills being higher in carbonate to the east. As was found in Saskatchewan (Christiansen, 1972b), carbonate content was the most useful index for till differentiation. Other properties, for example Ca, Mg and acid igneous sand grains, were closely associated with carbonate content. Additional variables were occasionally helpful in indicating till boundaries.

An oxidized zone directly below the upper till was observed in one-third of the drill-holes. It thus seems possible that sufficient time elapsed between the two glaciations for extensive weathering and/or soil development to occur. Oxidation of tills by groundwater action does occur but in the study area it appears to be sporadic, bearing little relation to compositionally-defined till boundaries. Such oxidation is usually associated with highly permeable zones within the surficial deposits. Additional evidence for the presence of a soil includes an oxidized zone between two tills, frequently in the absence of highly permeable material; the persistence of the zone at one stratigraphic position and the presence of a similar zone, interpreted as a weathered surface, in nearby areas of Saskatchewan (Christiansen, 1972b).

Underlying these two tills, but lacking any evidence of extensive weathering at its surface, the lower till has a low ( 1.5-5%) carbonate content and extends throughout the



map area. In the eastern part the electric logs suggest that it may be divided into two units (tills 3 and 4), although this distinction is not clear on the basis of its chemistry. Glacial or preglacial deposits may separate this till from the underlying bedrock shale or sandstone. The "Saskatchewan gravels and sands" (or "Saskatchewan gravels") are well described by Stalker (1968). They include many preglacial sand and gravel deposits in Alberta, but the important feature is that they are fluvial, lie unconformably on bedrock and can be shown to be nonglacial by their lack of igneous material derived from the Canadian Shield, their abundance of chert derived from the Cordillera and, except when located close to the Cordillera, by the absence of readily weatherable mineral species. Their presence in bedrock valleys is usually considered to confirm their preglacial origin. In this study, since interpretation is based exclusively on drill hole information, the presence of lithologically similar material at the base of a bedrock valley is considered to suggest a preglacial age for the valley. The presence of any Shield material in a clean sample of otherwise typically preglacial material would strongly suggest the later reworking of the preglacial deposits. While it is true that slumpage in the drill hole may introduce such material from above, a similarity of grain sizes for both the preglacial and shield material would suggest that they were, in fact, mixed prior to their being disturbed by drilling. In the plots of Appendix 4 the





suggested till units are marked in the right half of the central column. The upper till is labelled 1, the middle till 2, the upper member of the lower till is labelled 3 and the lower member 4.

## 7.2 Relation of Analyses to Till Units

The four till units suggested were distinguished on the basis of a wide range of properties, including their composition. At least some of the measured variables should therefore vary significantly between some of the tills. While this may be confirmed by visual inspection on a hole by hole basis for some variables, any general statistical test, runs into at least two problems with respect to sample location. Firstly, sample composition within a till varies appreciably and somewhat systematically over the map area. Carbonate content, for example, is likely to be lower in the western part of the map than in the east, whichever till is being considered, and hence it is unreasonable to compare directly samples from differing map locations. The second problem is that two of the units, the middle till and the lower part of the lower till, have only been identified in the eastern part of the area, thus biasing their compositional estimates if compared with units possessing better sample coverage.

In an attempt to minimize this problem, analysis of covariance was performed on each analyzed variable in turn, using the ANOVA program available in the SPSS package of





statistical programs (Nie et al., 1975). In this a one-way analysis of variance is performed to determine if the analyzed variable varies significantly between any of the four units, but at the same time the sample's location is incorporated in the analysis as two covariates (distances east and north in metres UTM coordinates). These covariates are first applied to explain as much of the variation as possible and the analysis of variance performed on the residual variance. Linear regression procedures are used on the raw data to remove variation due to the covariates, and thus more complex variations in composition with location are not considered. Nevertheless at least some of the bias due to location has been explained. These analyses are shown in Appendix 5. The variables were found to vary significantly at the 5% level between the four till units in all cases except the 20-40, 40-60, 60-140 mesh and total sand fractions.

If some of the variables differ significantly between till units, it is of interest to ask which variables differentiate which till units. Table 8 shows the grand mean for each variable and the mean for each till unit, corrected for location and expressed as deviations from the grand mean. These values are rank ordered with the unit expressed as a superscript from 1 to 4, as in Appendix 4. Duncan's "new multiple range test" (Steele and Torrie, 1960, Section 7.5) has been applied to them, and those means found not to be significantly different at the 5% level have been



Variable	Grand Mean	Unit Means			
Sand 10-20	3.13	<u>-.18<sup>3</sup></u>	<u>-.03<sup>1</sup></u>	<u>.27<sup>4</sup></u>	<u>.44<sup>2</sup></u>
Sand 20-40	5.35	<u>-.10<sup>2</sup></u>	<u>-.03<sup>1</sup></u>	<u>-.01<sup>3</sup></u>	<u>.41<sup>4</sup></u>
Sand 40-60	8.74	<u>-1.27<sup>2</sup></u>	<u>-.11<sup>3</sup></u>	<u>.01<sup>1</sup></u>	<u>.80<sup>4</sup></u>
Sand 60-140	14.73	<u>-1.49<sup>4</sup></u>	<u>-.64<sup>2</sup></u>	<u>-.06<sup>3</sup></u>	<u>.46<sup>1</sup></u>
Sand 140-230	5.74	<u>-.59<sup>2</sup></u>	<u>-.35<sup>4</sup></u>	<u>.01<sup>1</sup></u>	<u>.28<sup>3</sup></u>
Total Sand	37.7	<u>-1.37<sup>2</sup></u>	<u>-.74<sup>4</sup></u>	<u>-.06<sup>3</sup></u>	<u>.52<sup>1</sup></u>
Silt	51.37	<u>-1.42<sup>1</sup></u>	<u>.73<sup>3</sup></u>	<u>1.64<sup>4</sup></u>	<u>2.39<sup>2</sup></u>
Clay	10.93	<u>-1.02<sup>2</sup></u>	<u>-.89<sup>4</sup></u>	<u>-.64<sup>3</sup></u>	<u>.90<sup>1</sup></u>
CaCO <sub>3</sub>	6.78	<u>-1.60<sup>3</sup></u>	<u>-1.27<sup>4</sup></u>	<u>.13<sup>1</sup></u>	<u>4.78<sup>2</sup></u>
Local	7.07	<u>-3.76<sup>2</sup></u>	<u>-.16<sup>3</sup></u>	<u>.03<sup>4</sup></u>	<u>1.08<sup>1</sup></u>
Acid	59.18	<u>-2.82<sup>4</sup></u>	<u>-2.22<sup>3</sup></u>	<u>.98<sup>1</sup></u>	<u>4.20<sup>2</sup></u>
Basic	4.27	<u>-1.33<sup>4</sup></u>	<u>-.20<sup>3</sup></u>	<u>-.03<sup>1</sup></u>	<u>.89<sup>2</sup></u>
Carbonate	4.04	<u>-.62<sup>1</sup></u>	<u>.10<sup>3</sup></u>	<u>.84<sup>4</sup></u>	<u>1.57<sup>2</sup></u>
Quartz	25.44	<u>-2.90<sup>2</sup></u>	<u>-1.41<sup>1</sup></u>	<u>2.27<sup>4</sup></u>	<u>2.48<sup>3</sup></u>
Na	.31	<u>-.06<sup>4</sup></u>	<u>-.01<sup>1</sup></u>	<u>0.0<sup>3</sup></u>	<u>.06<sup>2</sup></u>
Mg	1.05	<u>-.18<sup>3</sup></u>	<u>-.18<sup>4</sup></u>	<u>.03<sup>1</sup></u>	<u>.50<sup>2</sup></u>
Al	7.63	<u>-.11<sup>2</sup></u>	<u>-.07<sup>1</sup></u>	<u>.07<sup>3</sup></u>	<u>.24<sup>4</sup></u>
Si	33.11	<u>-.96<sup>2</sup></u>	<u>.03<sup>4</sup></u>	<u>.05<sup>1</sup></u>	<u>.27<sup>3</sup></u>
K	2.04	<u>-.03<sup>3</sup></u>	<u>-.03<sup>4</sup></u>	<u>.01<sup>1</sup></u>	<u>.06<sup>2</sup></u>
Ca	2.91	<u>-.35<sup>3</sup></u>	<u>-.19<sup>4</sup></u>	<u>.04<sup>1</sup></u>	<u>.93<sup>2</sup></u>
Ti	0.44	<u>-.02<sup>2</sup></u>	<u>-.01<sup>1</sup></u>	<u>.02<sup>3</sup></u>	<u>.02<sup>4</sup></u>
Fe	3.82	<u>-.05<sup>1</sup></u>	<u>0.0<sup>2</sup></u>	<u>.06<sup>3</sup></u>	<u>.08<sup>4</sup></u>

Table 8 Variation of analyses between till units. Column 2 shows the grand mean for each variable and columns 4 - 7 show the values for each unit, expressed as deviations from the grand mean. Superscripts indicate the till unit. Underlined means do not differ from each other significantly.



underlined. Table 9 summarizes which variables may satisfactorily distinguish between particular tills in this study. Thus a variable plotted in a particular cell has a significantly higher value in the till forming the column than in the till forming the row. On the right of Table 9 are shown those variables whose mean value for a till differs significantly from all the other tills, either by being high or low. Those variables in parentheses are second highest or lowest in Table 8. It can be seen that all till units may be distinguished on the basis of composition, but the lower member of the lower till only differs from the upper member by having a somewhat coarser sand texture and a higher Al content.

### 7.3 Relationships between Analyzed Variables

Clearly the behaviour of some of the variables in Table 9 is similar to the behaviour of other variables. It would be of value for future work to determine which analyzed variables group together and hence which ones are redundant. Cluster and factor analysis are suitable techniques for examining this problem.

Cluster analysis (Hartigan, 1975, Sneath and Sokal, 1973) is a method of partitioning an observed sample population into different classes and producing hierarchical classifications of these. In this case the 22 analyzed variables themselves were considered the samples to be classified. Several different forms of cluster analysis





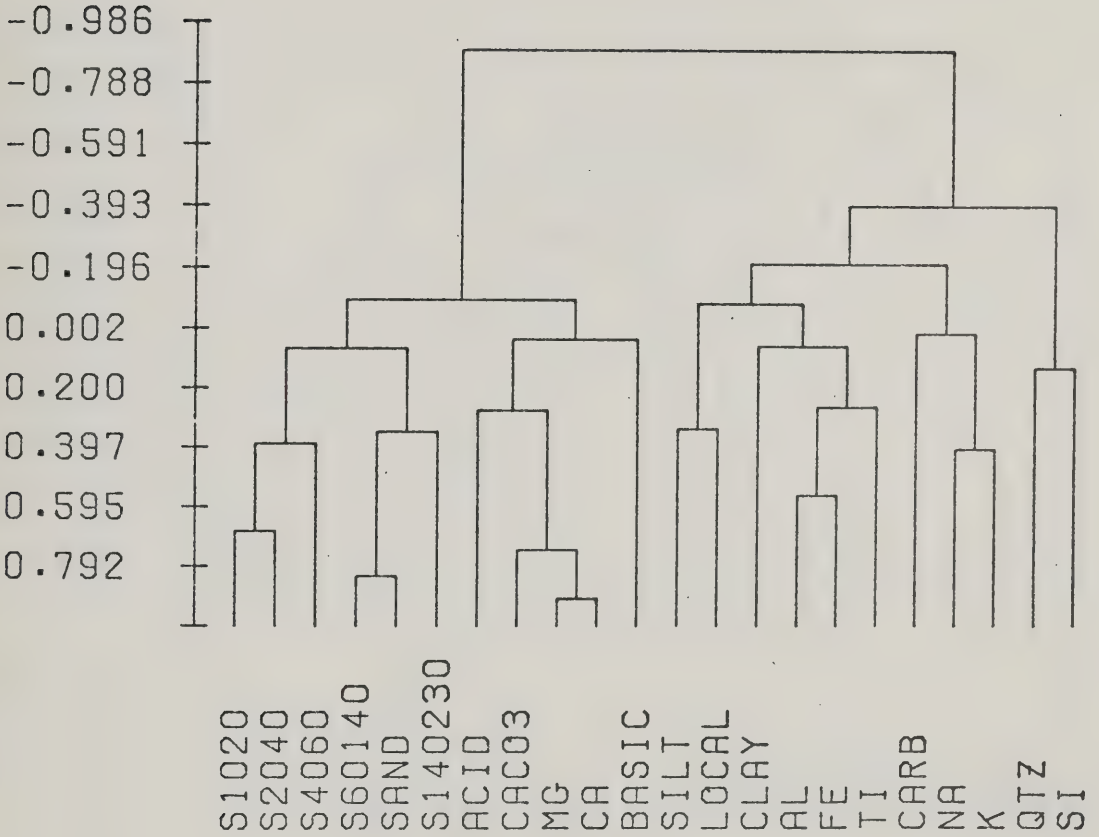
		Unit with Higher Value				Variables Diagnostic for each Unit	
		1	2	3	4	Low	High
Unit with Lower Value	1		S1020 Silt CaCO3 Acid Basic Carb Na Mg Ca K	Silt Carb Qtz Al Ti Fe	Silt Carb Qtz Al Ti Fe	Silt Local Carb (Qtz)	Clay (CaCO3) (Acid) (Mg) (Ca)
	2	S60140 S140230 Clay Local Qtz Si		S140230 Local Qtz Al Si Ti	S4060 Local Qtz Al Si Ti	Qtz Si	CaCO3 Acid Basic Mg K Ca
	3	Clay CaCO3 Acid Mg K Ca	S1020 CaCO3 Acid Basic Carb Mg Ca K		S1020 S4060 Al		(Al)
	4	S60140 Clay CaCO3 Acid Mg Ca	CaCO3 Acid Basic Na Mg K Ca	S60140 S140230			Al

Table 9 Analyzed variables that differentiate between till units. As an example, row 1, column 4 in the left hand table indicates that Fe is significantly higher in unit 4 than in unit 1. The right hand table indicates those variables that differentiate a till unit from all others. Variables in parentheses are second highest or lowest (see Table 8).



available in the CLUSTAN package (Wishart, 1975) were performed using the correlation matrix given in Appendix 6 as the information describing the relations between variables. Figs. 28 and 29 show dendrograms produced by hierarchical clustering, in which initially each variable forms a separate cluster and clusters are grouped in order of greatest similarity. Similarity between groups may be defined in several different ways. Fig. 28 shows the result when group similarity is defined as the lowest similarity between any pairs of points, one from each cluster. Fig. 29 shows the result when the average, rather than the lowest, pair similarity is used. Fig. 30 shows the results of using the "dendrite" procedure in which the minimum spanning tree, or shortest dendrite connecting all points, (Wishart, 1975, p. 67) is broken into the desired number of clusters by eliminating the required number of connections. The linkages removed are chosen on the basis of minimizing an error sum term. A final dendrogram (Fig. 31) is produced by the "mode" procedure in which a sphere of a chosen radius is considered to enclose each data point, and clusters are defined as all those points whose spheres form a common bounding surface. The method of introduction of points and updating the sphere radius is such that at the start and finish only one cluster is present, and it has been suggested that the number of clusters present prior to the first fusion of groups represents the lowest "natural" level of classification, in this case three clusters.

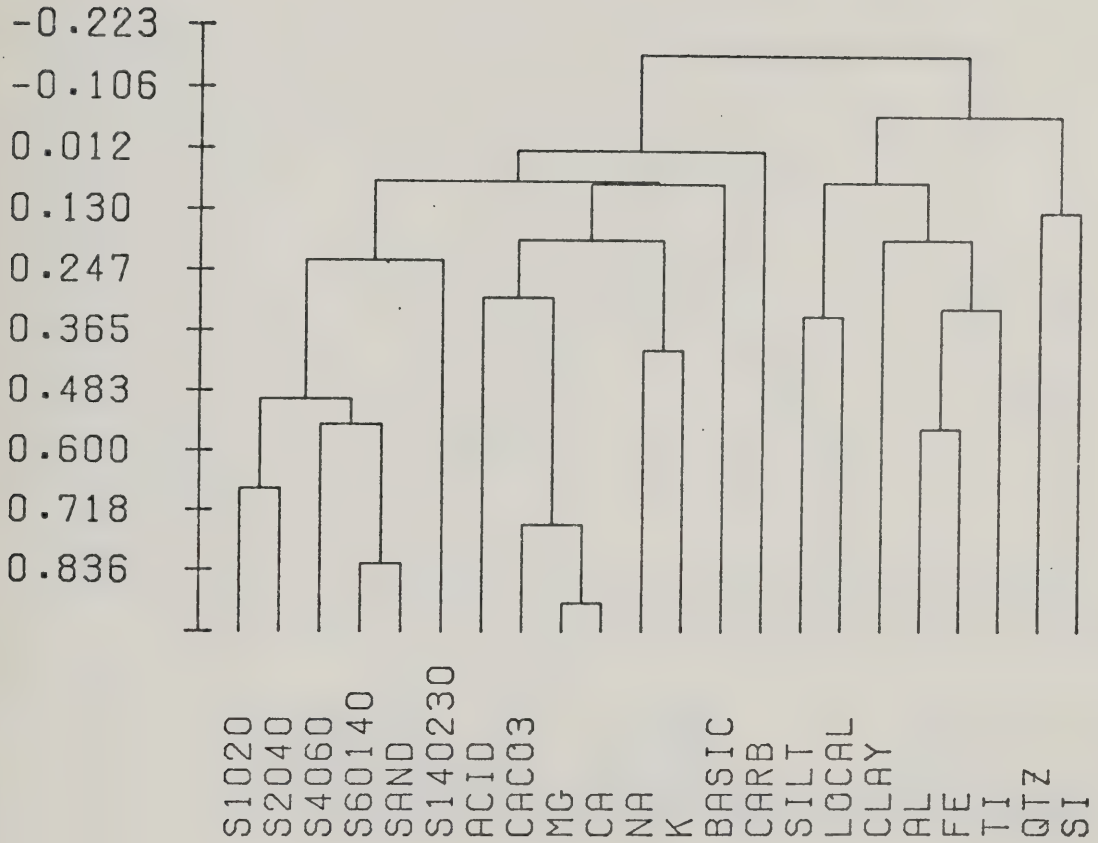




HIERARCHICAL CLUSTERING (FURTHEST NEIGHBOUR)

Fig. 28 Dendrogram showing the relationships between the 22 analyzed variables. Ordinate values represent the product - moment correlation coefficient.



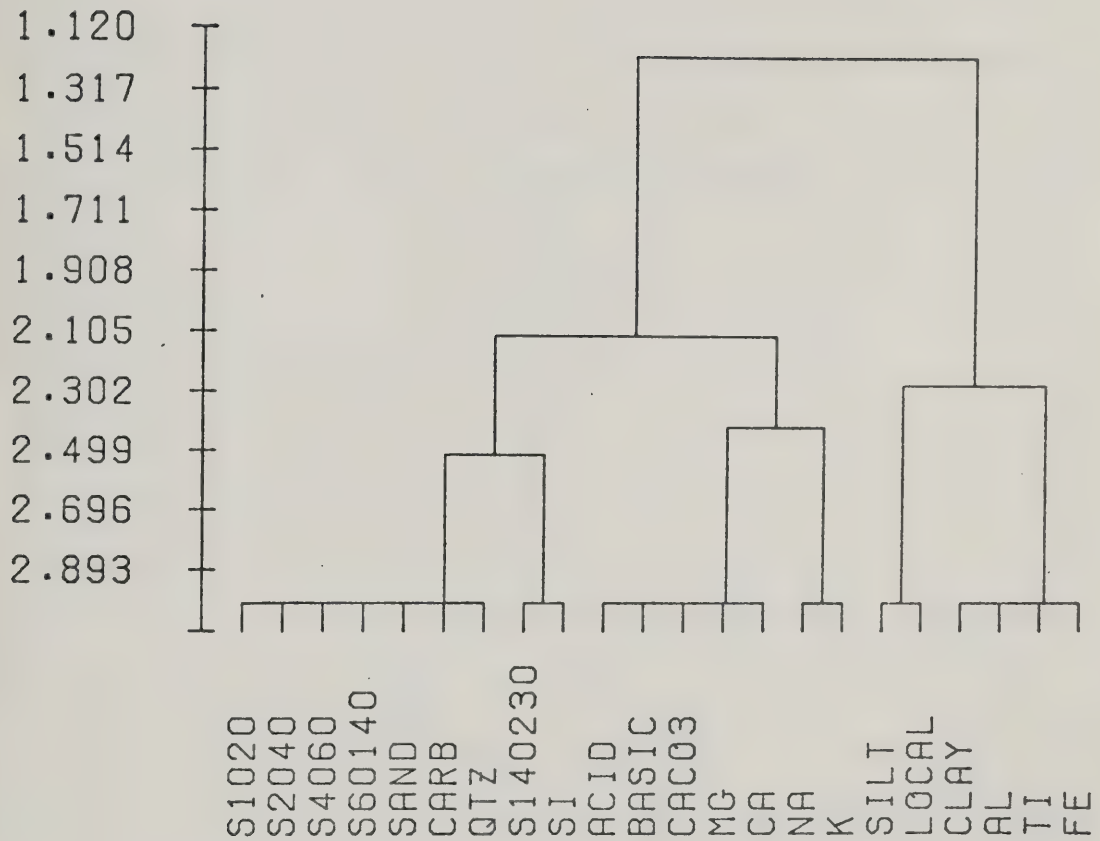


HIERARCHICAL CLUSTERING (UNWEIGHTED PAIR GROUP)

Fig. 29 Dendrogram showing the relationships between the 22 analyzed variables. Ordinate values represent the product - moment correlation coefficient.







### DENDRITE ANALYSIS

Fig. 30 Dendrogram showing the relationships between the 22 analyzed variables. Ordinate values represent the within - groups error sum of squares.



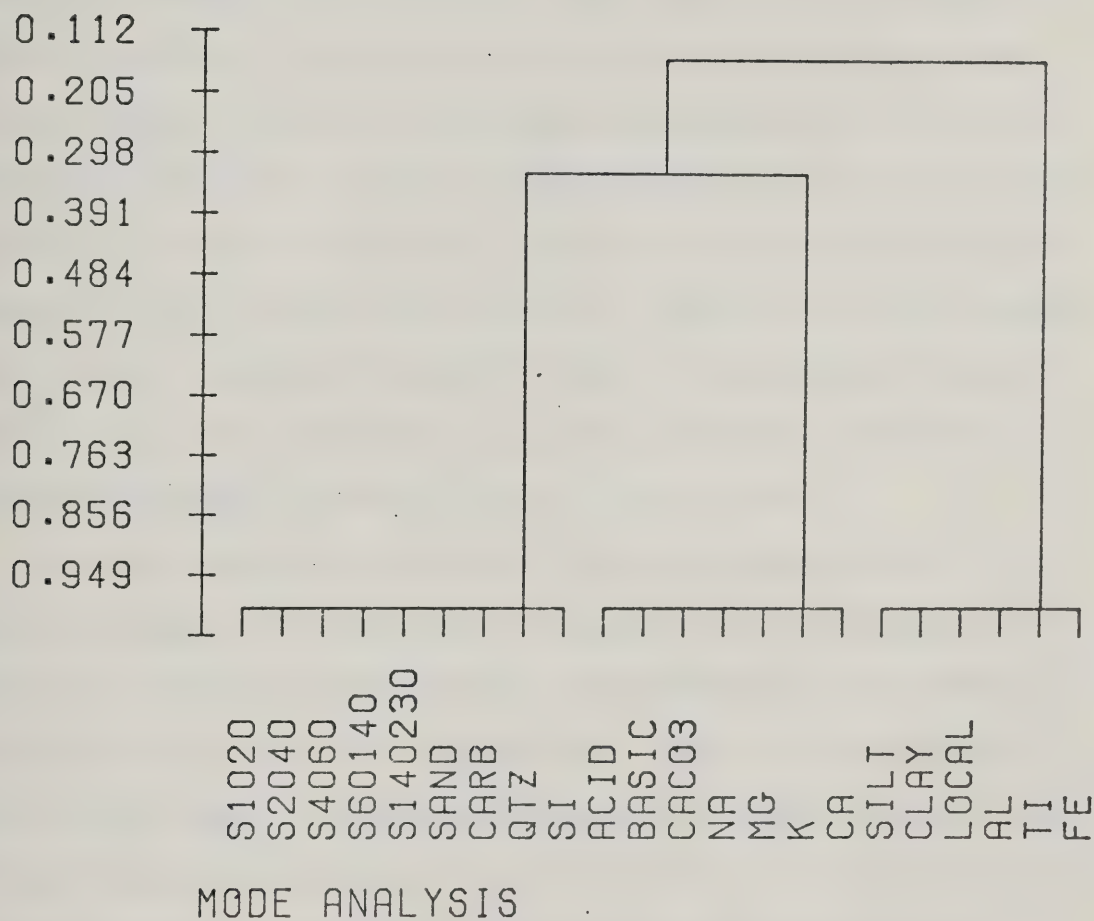


Fig. 31 Dendrogram showing the relationships between the 22 analyzed variables. Ordinate values represent the product - moment correlation coefficient.



The cluster methods illustrated all agree in general on these three major groups. Group 1 consists of all sand sizes. Carbonate and quartz grains and Si content may associate weakly with the sand sizes, but not consistently between methods. Group 2 consists of components that could be considered "erratic" in origin, possible sub- groups 2a containing Na and K and 2b containing acid and basic grains, calcium carbonate equivalent, Mg and Ca. The association of groups 2a and 2b with each other is agreed on by all except the furthest neighbour method. Group 3 consists of "local" constituents -3a consisting of silt percentage and local grains, 3b consisting of clay, Al, Ti and Fe.

Factor analysis (Cattell, 1965a,b, Harman, 1967, Kaiser, 1970) is a technique for examining a correlation matrix to determine whether its dimensions may be reduced without serious loss of information. This is achieved by creating factors or components which are linear combinations of the observed variables. The first step is to define a set of uncorrelated components that explain as much of the variation of the data as possible. The second step is to determine if any of these components are of insignificant value in explaining the data variation, and eliminating them. The third step is to rotate these components until they are associated in some fashion with groups of the original variables. Appendix 6 contains the results of factor analysis of the 22 analyzed variables, using the principal component approach and varimax rotation.





Calculations were performed using the SPSS package (Nie et al., 1975). Although seven eigenvalues are greater than unity, a scree test (Cattell, 1966) on these suggests that a maximum of four are meaningful. Component 1 associates strongly with the sand size fractions, and negatively with the silt fraction. Component 2 associates strongly with calcium carbonate equivalent, Ca and Mg, and moderately with 10-20 mesh sand, acid and basic grains, Na and K. There is strong negative association with Si and, to a lesser extent, quartz grains. It can be related to erratic components of the till, especially the Devonian carbonates and, perhaps to a lesser extent, the igneous Precambrian shield to the northeast. Component 3 associates strongly with clay, Al, Ti and Fe, suggesting a relationship with the local shales. Component 4 associates strongly and positively with Na, K and quartz grains but moderately and negatively with Si, suggesting a possible relation to quartz grains and feldspars from the Precambrian shield to the northeast. Component 5 associates fairly strongly with carbonate and quartz grains, Component 6 with the fine sand fractions and Component 7 associates positively with acid and basic grains but strongly negatively with local grains.

A comparison with the cluster analysis results suggests that components 1,2 and 3 correspond well with cluster groups 1,2 and 3, but that there may well be a distinction between cluster groups 2a and 2b based on the source of the erratic material.



Analyzed variables may thus be classified into textural, erratic and local groups, implying that in further work fewer variables need be examined. In addition, the use of factor analysis techniques permits the derivation of composite component scores. Thus the Component 2 score for a sample may better represent the erratic component of the till than an individual variable. While such scores were not used in the visual estimation of the till unit contacts (Appendix 4) they have been used in the examination of compositional variation within a till unit.

#### 7.4 Choice of Variables for Till Differentiation

Examination of the plots of each variable against depth suggests a question: "Which variables can one use in order to distinguish different till units within a drill hole?" On the basis of cluster and factor analysis the most useful variables are those expressing the local/ erratic proportions in the tills. Ca, Mg and calcium carbonate equivalent are obvious choices that estimate the content of erratic material. Clay content, even with the poor accuracy of analysis in this study due to long sample storage, estimates the local material content fairly well, but not as well as Al or Fe. Total sand content, although not diagnostic of any particular till units, often is useful to mark breaks in depositional continuity. Acid igneous grains are also occasionally helpful. However, for identifying till units as continuous between drill holes several miles apart,



variation in the erratic component is the most useful. In practical terms, percent total sand and percent calcium carbonate equivalent are the variables most commonly used to distinguish till layers, and calcium carbonate equivalent alone is frequently adequate to correlate these units between drill holes. A good example of clearly distinguished tills can be seen in drill hole 714E in Appendix 4. It is also evident that the bulk chemistry of the till matrix (less than 63 microns in this study) exhibits a far greater degree of homogeneity within the till units than would be expected from examination of the coarser fractions.

#### 7.5 Compositional Variation within Till Units

On the basis of the suggested till units and groupings of analyzed variables, how should the compositional variation in the tills be described? Apart from vertical variation in composition between till units there is vertical variation within till units at a given location. In addition there is the lateral variation of a particular till. Boulton and Dent (1974) describe grain size variation due to shearing and weathering in the top metre or so of an exposed lodgement till, but in the current study the variation under consideration is on a larger scale. Appendix 7 illustrates a property of the lateral variations of the analyzed components of the four till units -- it may be very large. These maps were produced using the triangulation techniques described earlier. If lateral variation is indeed





larger than vertical variation on a regional basis, then a natural question concerns the local area over which lateral consistency is observed, since some lateral consistency must exist in order to permit the observation of vertical till homogeneity. A final question concerns the similarity of behaviour, or lack of it, between the different analyzed components of the till.

### 7.6 Vertical Variation in Till Composition

Hole number 714E was chosen to illustrate the vertical variation in till composition, since three well-sampled till units were identified in it. The middle till (unit 2 in the analysis of variance tables of Appendix 8) was sampled 19 times, the upper member of the lower till (unit 3) was sampled 13 times, and the lower member of the lower till (unit 4) eight times. Appendix 8 contains the results of a one-way analysis of variance performed on each of the 22 analyzed variables, with the three till units as the classifying categories. The main question is whether the between-units variation is significantly greater than the within-units variation, in which case that particular analyzed variable may be used to distinguish between at least two of the till units found in this drill hole. Appendix 8 shows that this is the case for the 10 to 20 mesh sand, all grain counts except the basic igneous grains, calcium carbonate equivalent, and all the bulk chemical analyses. Calcium carbonate equivalent and calcium and





magnesium content showed extremely high levels of significance.

In addition to the analysis of variance tables, Appendix 8 shows the mean, range and 95% confidence interval of the mean for each of the till units. From these it can be determined which tills may be distinguished using a particular analysis. The middle till (unit 2) may be distinguished quite well from both the lower tills by the 10 to 20 mesh sand, local, acid igneous and quartz grains, calcium carbonate equivalent and all bulk chemistry analyses. The upper and lower members of the lower till (units 3 and 4) may be distinguished by calcium carbonate equivalent, Mg, Si, Ca and Ti concentrations. In addition, erratic components tend to increase in concentration, and local components decrease, with decreasing age of the till unit. This is a variation between units, rather than a simple variation with depth, since for many analyses the between-units variance was much greater than that within units. Hole 714E shows compositional differences between tills better than most of the drill holes. It does however, illustrate the high possible homogeneity of the composition of a till unit throughout its thickness, especially in the fine fraction.



### 7.7 Local Depositional Processes

In contrast, the lateral homogeneity of till composition is far less clear, and the composition maps of Appendix 7 may show appreciable fluctuation in composition over the study area. This suggests that till forming processes were local in effect but stable throughout the period of deposition of at least the basal part of a till unit, although distinction of basal from ablation till has not been attempted in this project. Sufficient information is not available to suggest a depositional mechanism that, while only producing the same effects over short distances laterally, maintained these effects during most of the depositional period of the till. It appears unlikely that such a process would remain locally stable over extended periods of time while the flow conditions of the ice sheet changed in response to climatic and other conditions. It appears much more likely that the actual deposition of till only occurred as a single relatively short-lived phenomenon in the life-cycle of a continental ice-sheet. Under these conditions the bulk of the till unit - up to 100 ft (30m) in the case of hole 714E, and greater in some others - would have been deposited under one, perhaps very short-lived, set of local flow conditions. While deposition may have occurred more than once, under different local conditions, there is little evidence for this in the upper and middle till units, although this is a possible explanation for the weak but sometimes measurable differences between the upper and lower



members of the lower till.

### 7.8 Lateral Variation in Till Composition

If depositional processes were involved that remained constant over only limited lateral extent it is important to examine the extent of those possibly local processes. An attempt was made to investigate this by using a suitable analysis of variance design. In addition, since some till constituents were described as being erratic components, it was necessary to determine whether they or any other analyzed variables could be considered to vary significantly between different parts of the study area.

### 7.9 Sampling and Statistical Model

In order to compare local and regional lateral compositional variation it was necessary to select several distinct areas of the map each containing several drill holes. In addition, if between-units variation was to be examined these drill holes should penetrate several tills, and if vertical variation is of interest, several samples should be available from each unit. Unfortunately, since computational methods become extremely complex for suitable analysis of variance designs if some categories contain no samples, the choice of test data was very limited. Consequently three areas were selected, a western one using holes 435E, 436E, 437E and 758E. A central area containing holes 456E, 498E and 709E; and an eastern area using holes





457E, 458E and 765E. Since the middle till and the lower member of the lower till appear to be absent from the west of the area, only the upper till and the upper member of the lower till were used. This was perhaps unfortunate since the middle till was compositionally the most distinctive. Finally, due to the low number of samples in some drill holes only two samples were used from each unit in each drill hole, selected from near the top and bottom of the unit.

The analysis of variance design used was a model described by Winer (1962, p. 334, Table 7.3-10). The model would have been a simple factorial design were it not for the fact that a particular drill hole clearly cannot be repeated in each of the three areas. Hence drill holes are "nested within" areas and are treated as replicates. However, the unit and position within the unit of each sample must still be accounted for and are "crossed" with each other and with the drill holes, since a sample may be found for each combination of drill hole, unit and sample position. This model is most easily defined as a single factor model (area) with replicates (drill holes) and repeated measures. These repeated measures therefore consist of the four combinations of unit and sample position. It is a property of this model (and of the data) that interaction terms involving drill holes cannot be tested and must be pooled with other terms to form the relevant error sums of squares. Thus the sum of squares due to area is tested



against a pooled error sum of squares consisting of the sum of squares due to drill holes plus the interaction between drill holes and areas. This interaction has no conceptual meaning since drill holes are nested within areas. Similarly the significance of units, and the interaction between areas and units, are tested against the interaction between units and drill holes plus the interaction between units, drill holes and areas. Sample position and (area x position) interaction are tested against the (drill hole x position) plus (drill hole x position x area) interactions. Finally, the (unit x position) and (area x unit x position) sums of squares may be tested against the pooled (unit x position x drill hole) plus (unit x position x drill hole x area) sums of squares.

Appendix 9 shows the analysis of variance results for each of the 22 variables and Table 10 summarizes the significance of the relevant F, or variance ratio, tests. A particular concern is the relation between local and regional variation for each component of the till. Those analyses differing significantly between regions of the map as opposed to within regions were acid igneous grains, calcium carbonate equivalent and Ca content at the 5% level and Na at the 10% level, with Mg approaching 10% significance. Thus, although the variance analysis was performed on fewer samples than would have been desirable, it appears that only erratic constituents of the till vary more on a regional basis than a local one within a specified



	Area	Unit	Unit x Area	Position	Position x Area	Unit x Position	Unit x Position x Area
Sand 10-20	.142	.144	<u>.012</u>	.112	.349	.617	.183
Sand 20-40	.542	.106	<u>.023</u>	<u>.088</u>	.430	.390	.490
Sand 40-60	.274	<u>.009</u>	<u>.001</u>	.888	.611	<u>.036</u>	.101
Sand 60-140	.641	.366	<u>.052</u>	.652	.387	.692	.109
Sand 140-230	.426	.346	.988	.563	.372	.988	.467
Total Sand	.337	.135	<u>.004</u>	.238	.254	.331	<u>.050</u>
Silt	.515	.164	<u>.001</u>	<u>.067</u>	.200	.605	.584
Clay	.529	<u>.028</u>	.126	.553	.125	.452	<u>.032</u>
Local	.218	.738	.175	.743	.814	.839	.445
Acid	<u>.037</u>	.870	.223	.202	.555	.554	.707
Basic	.535	.761	.541	.449	.320	.806	.609
Carbonate	.990	.857	.317	.439	.566	<u>.033</u>	.616
Quartz	.300	.611	.221	.593	.994	.521	.895
CaCO <sub>3</sub>	<u>.001</u>	.191	.338	.835	.605	.115	.147
Na	<u>.059</u>	.906	.704	<u>.011</u>	.199	.482	.696
Mg	.118	.105	.775	<u>.099</u>	.819	.145	.113
Al	.680	.922	.321	.737	.279	.856	.113
Si	.156	.643	.723	.191	.899	.562	.844
K	.197	.433	.888	.568	.849	.888	.980
Ca	<u>.020</u>	.578	.634	.221	.519	.418	.281
Ti	.248	.322	.194	.124	.531	.394	.186
Fe	.854	.429	.502	.107	.334	.729	.512

Table 10. Significance levels of the factors and interactions of Appendix 9. Levels of 10% or better are underlined.



till unit. Appendix 7 shows the lateral variation in each till unit for grain size, grain count, calcium carbonate equivalent and bulk chemistry.

The till components that do vary significantly between regions of the map are those that correspond strongly with Component 2 in the factor analysis described in Section 7.3. It was suggested that this component represented erratic material introduced into the map area by the moving ice. As can be seen from the maps of Appendix 7 these same components, despite any local variations, decrease generally towards the southwest. Besides maps of calcium carbonate equivalent for each till unit, Figs. 32 to 35 contain maps for each till unit of the Component 2 scores for all samples, calculated from the factor-score coefficient matrix of Appendix 6. The score for each sample is obtained by normalizing each analysis value (by subtracting the mean and dividing by the standard deviation for that analysis), multiplying it by the Component 2 score coefficient for that analysis, and then summing these results over all analyses. In addition Figs. 36 to 39 show maps of the ratio of the concentrations of erratic cations (Na, Mg, K, Ca) over local cations (Al, Si, Ti, and Fe) that are probably derived mainly from the local sandstones and shales. Calcium carbonate equivalent, Component 2 scores and erratic ratios tend to behave in a similar fashion for a particular till unit.

Dreimanis and Vagners (1969, 1971) examined the





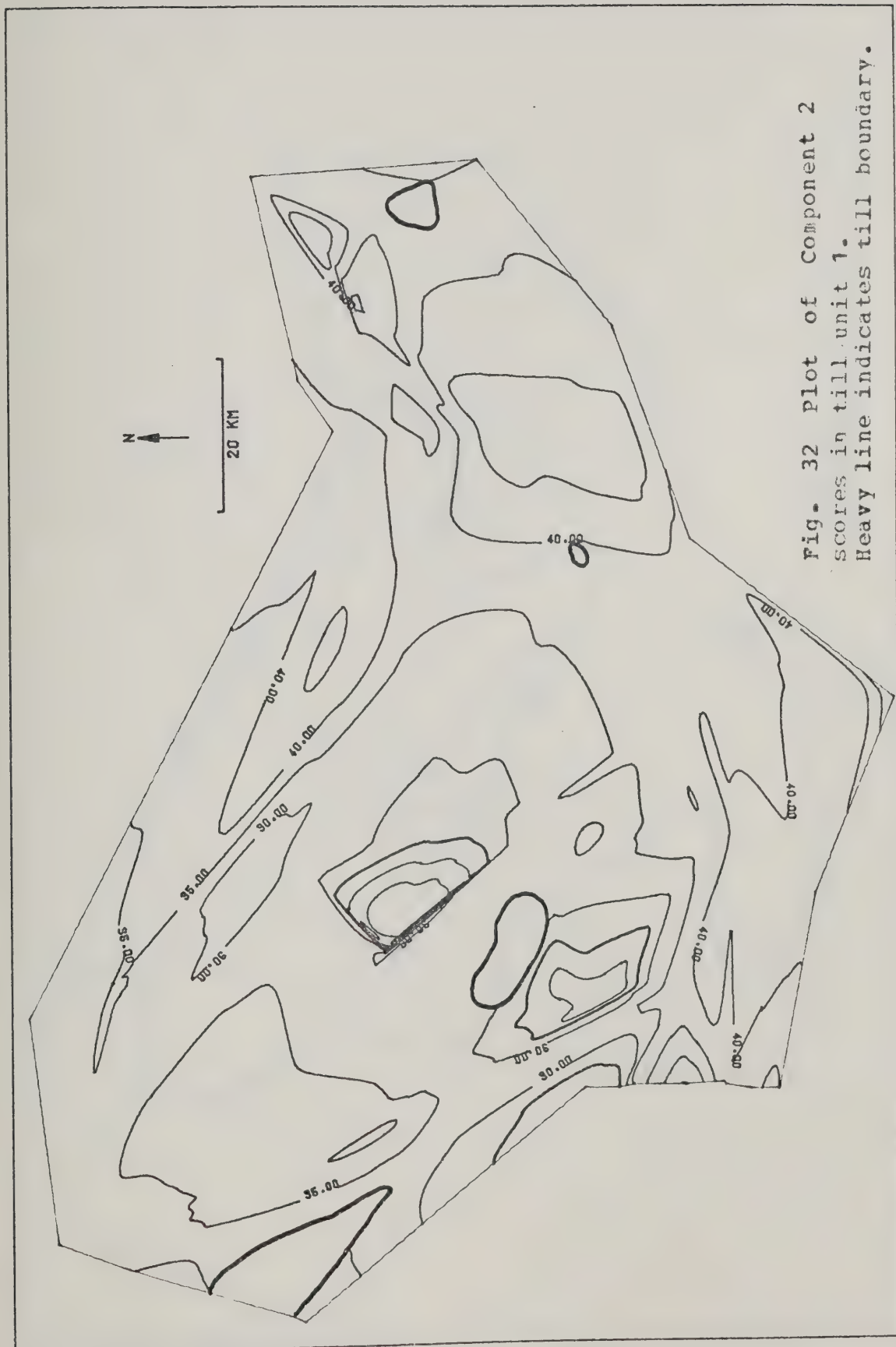


Fig. 32 Plot of Component 2  
scores in till unit 1.  
Heavy line indicates till boundary.



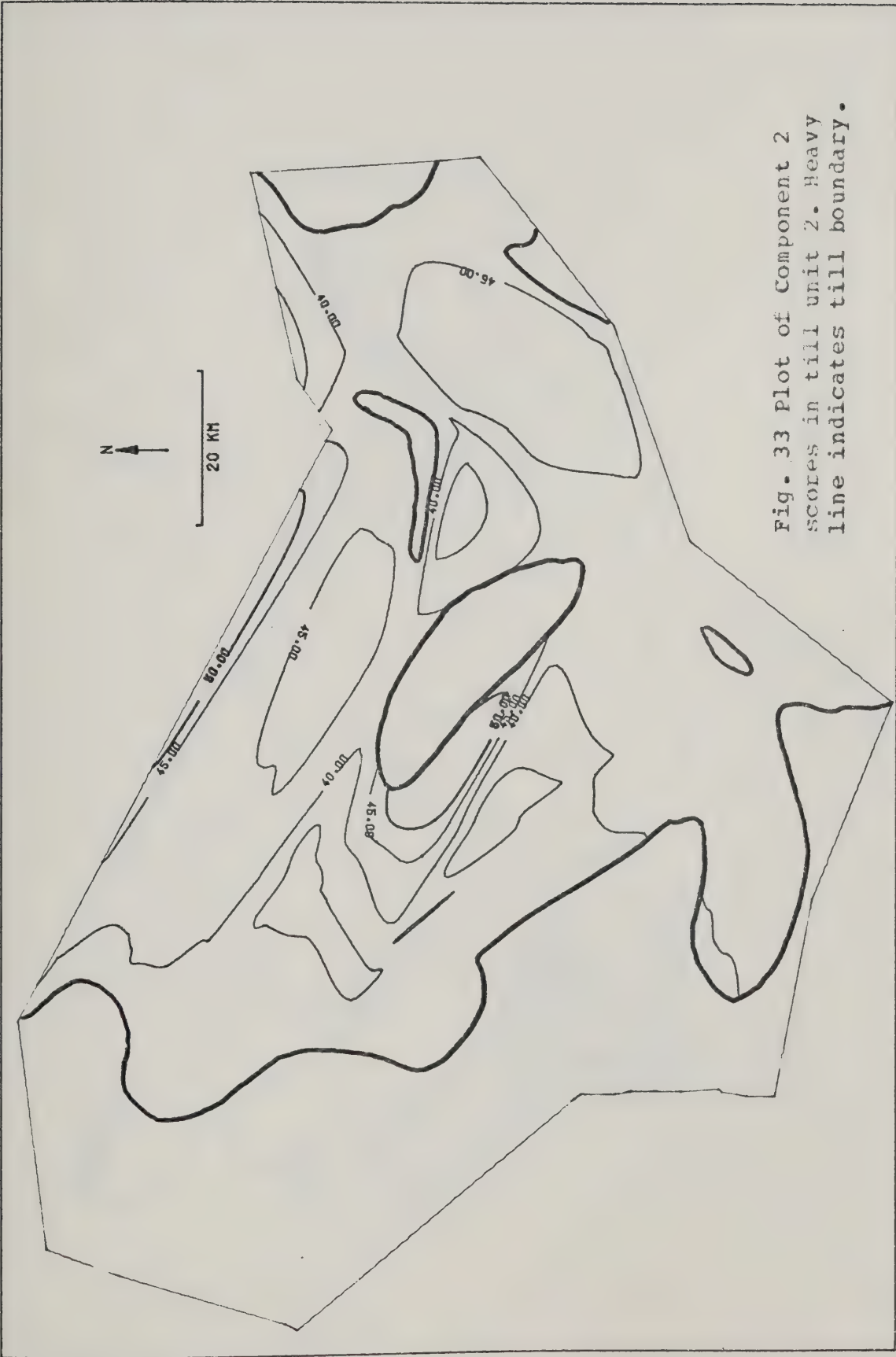


Fig. 33 Plot of Component 2 scores in till unit 2. Heavy line indicates till boundary.



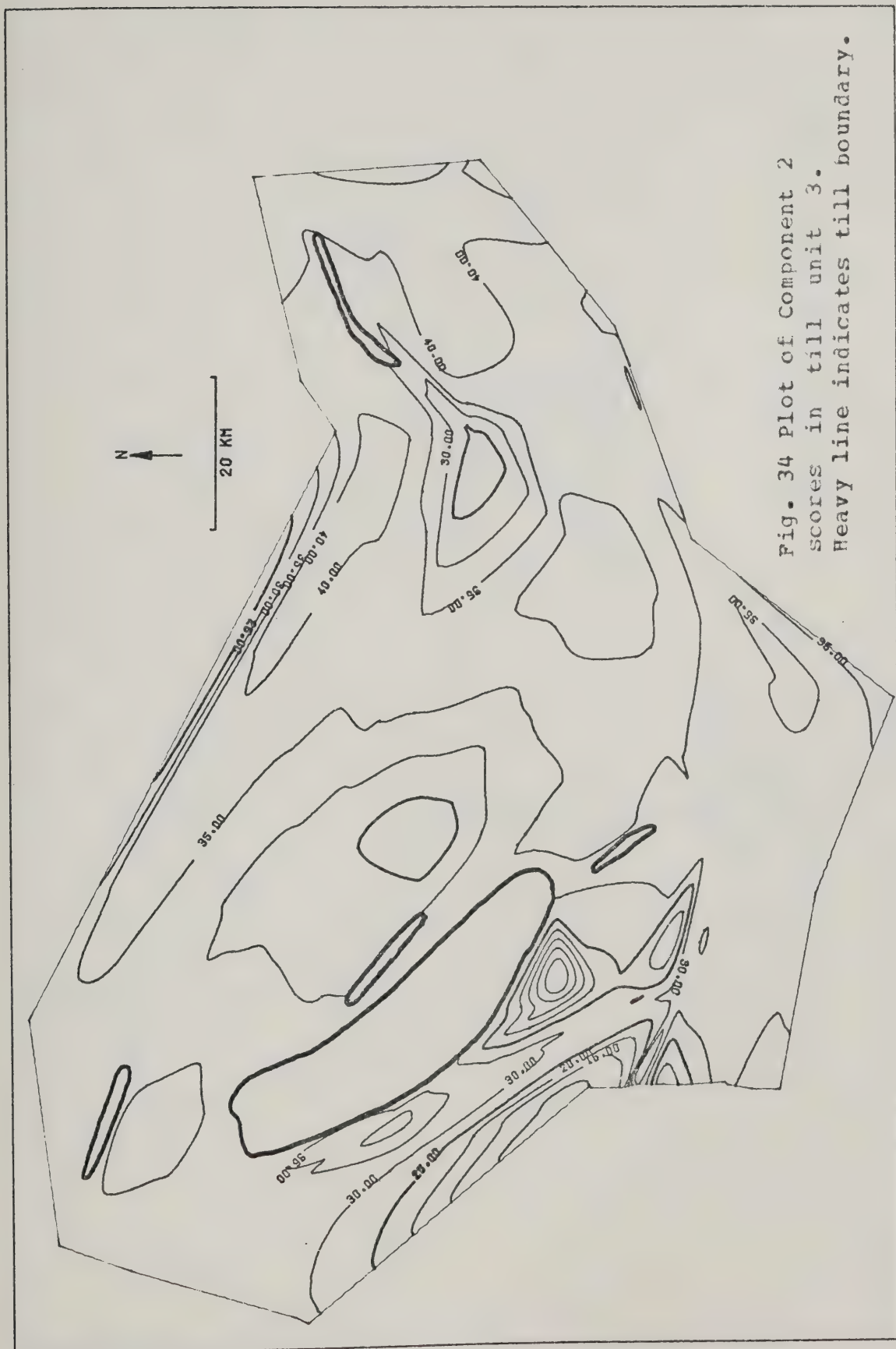


Fig. 34 Plot of Component 2 scores in till unit 3. Heavy line indicates till boundary.





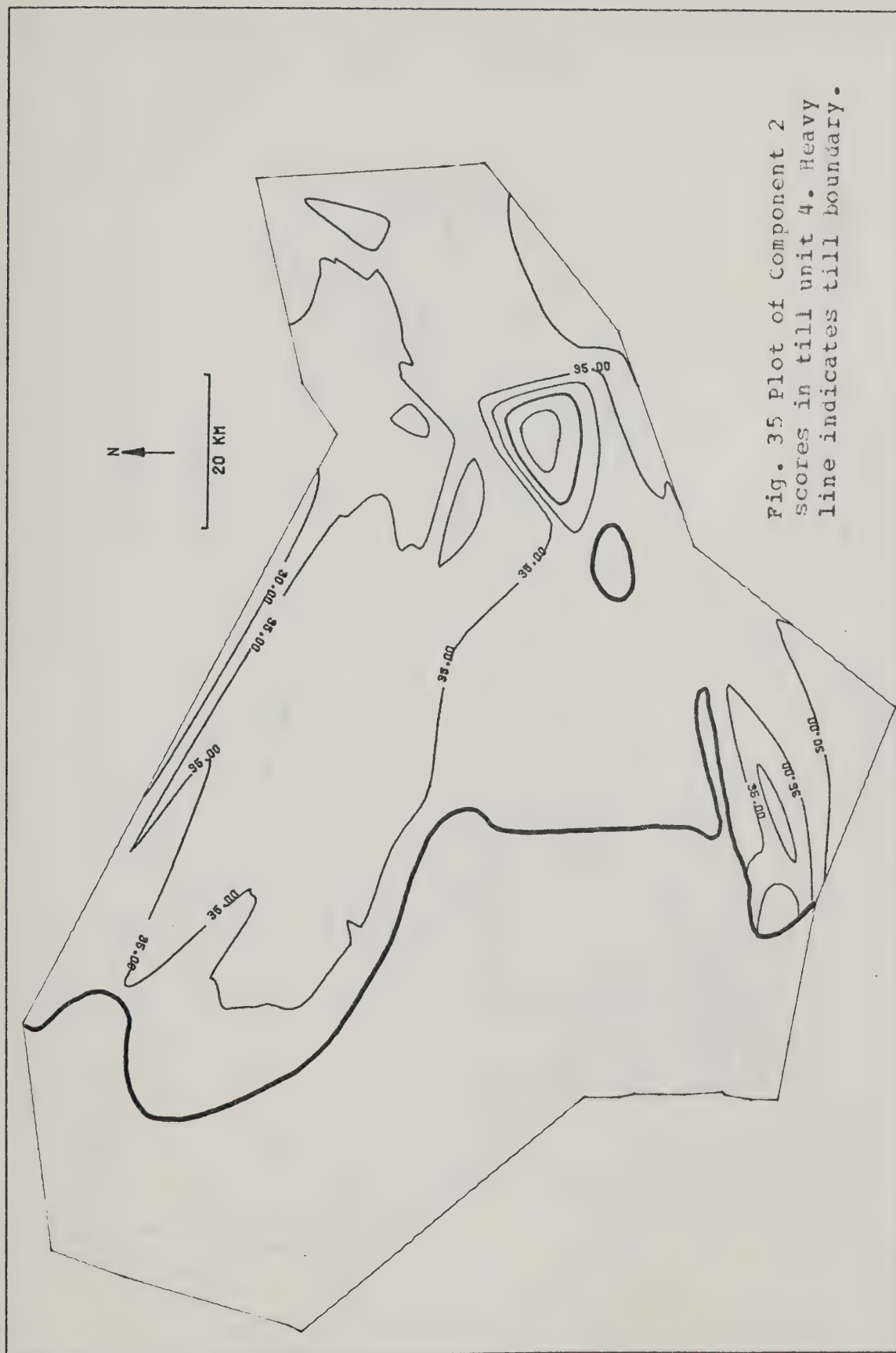
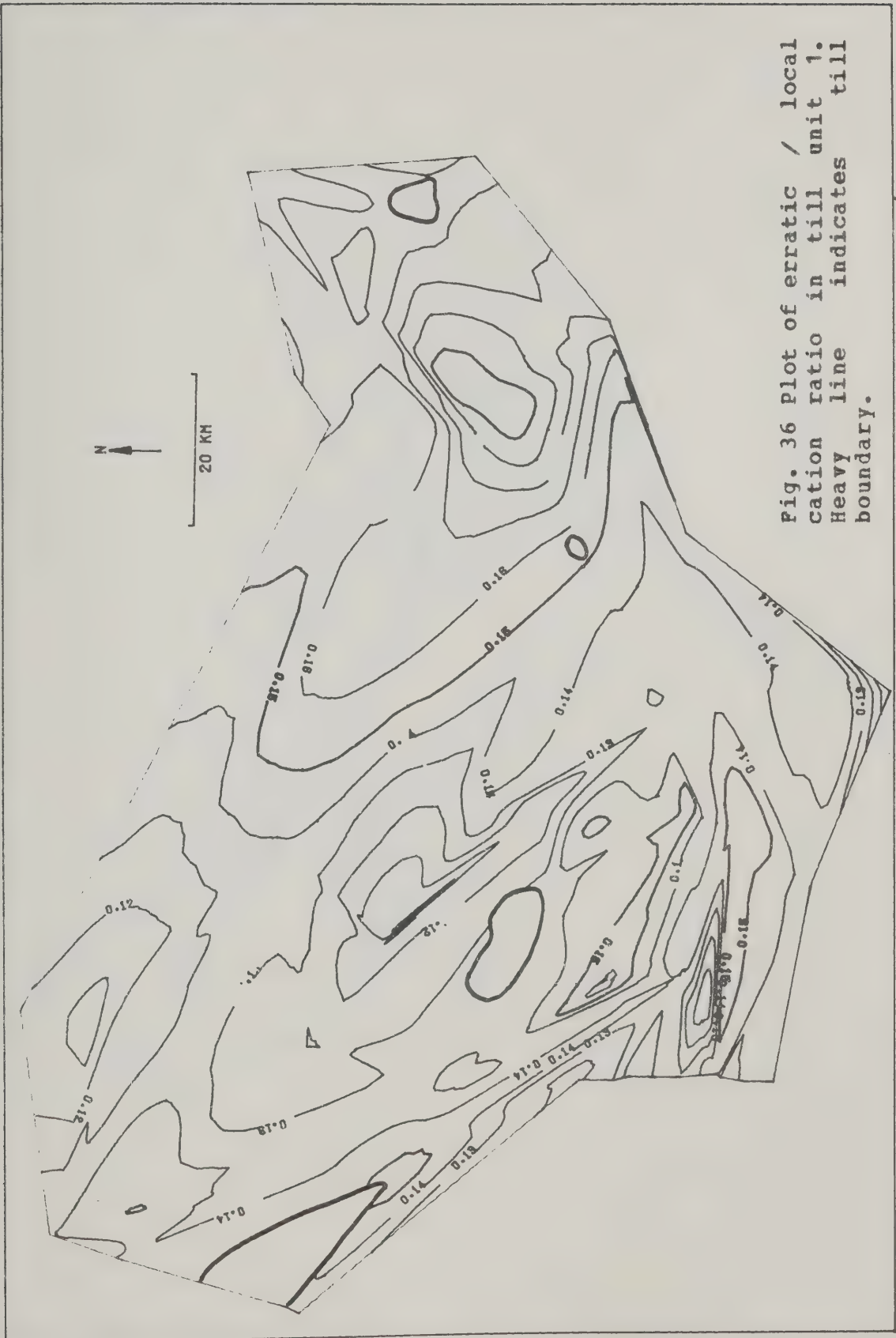


Fig. 35 Plot of Component 2 scores in till unit 4. Heavy line indicates till boundary.







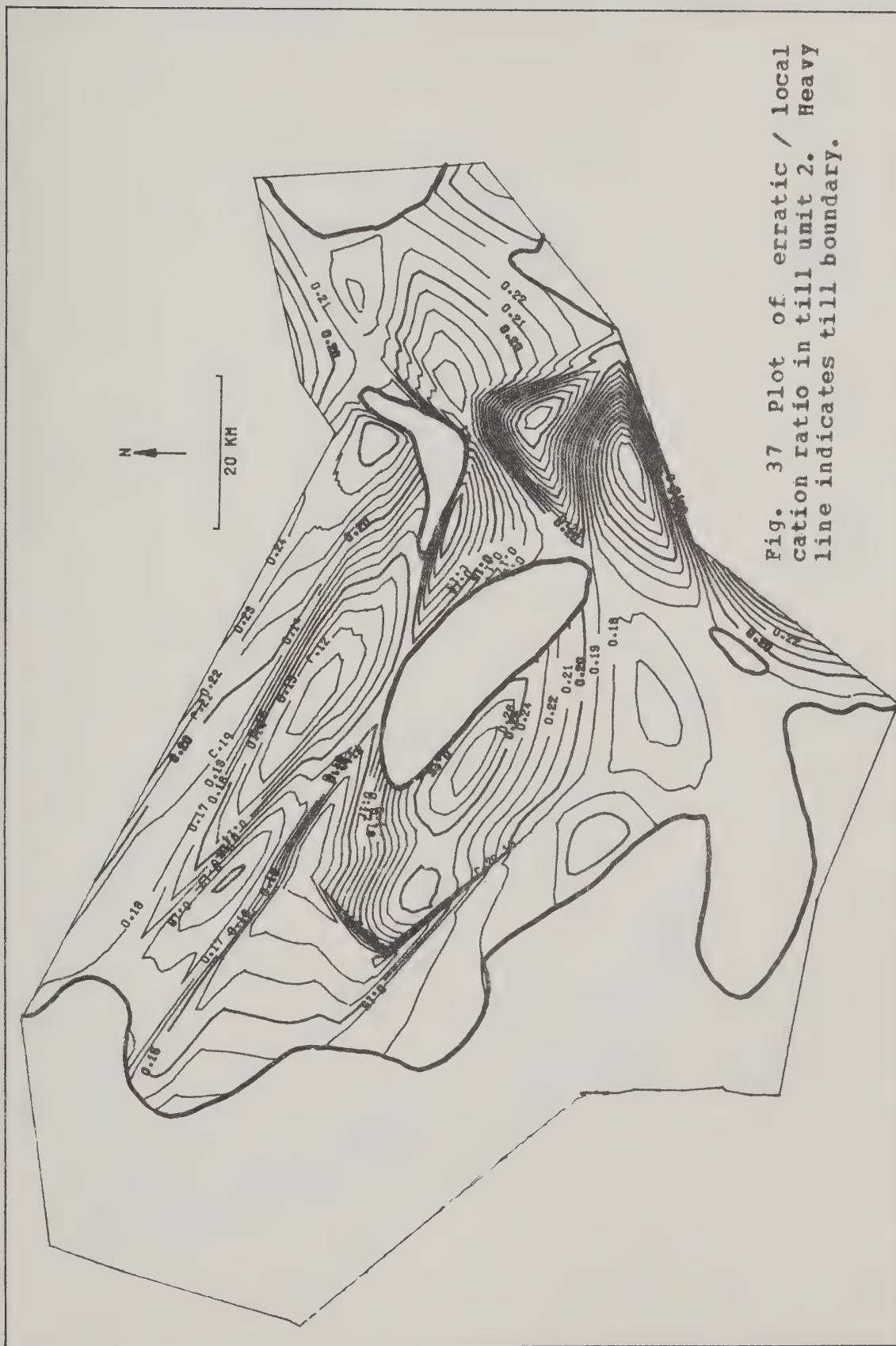


Fig. 37 Plot of erratic / local cation ratio in till unit 2. Heavy line indicates till boundary.



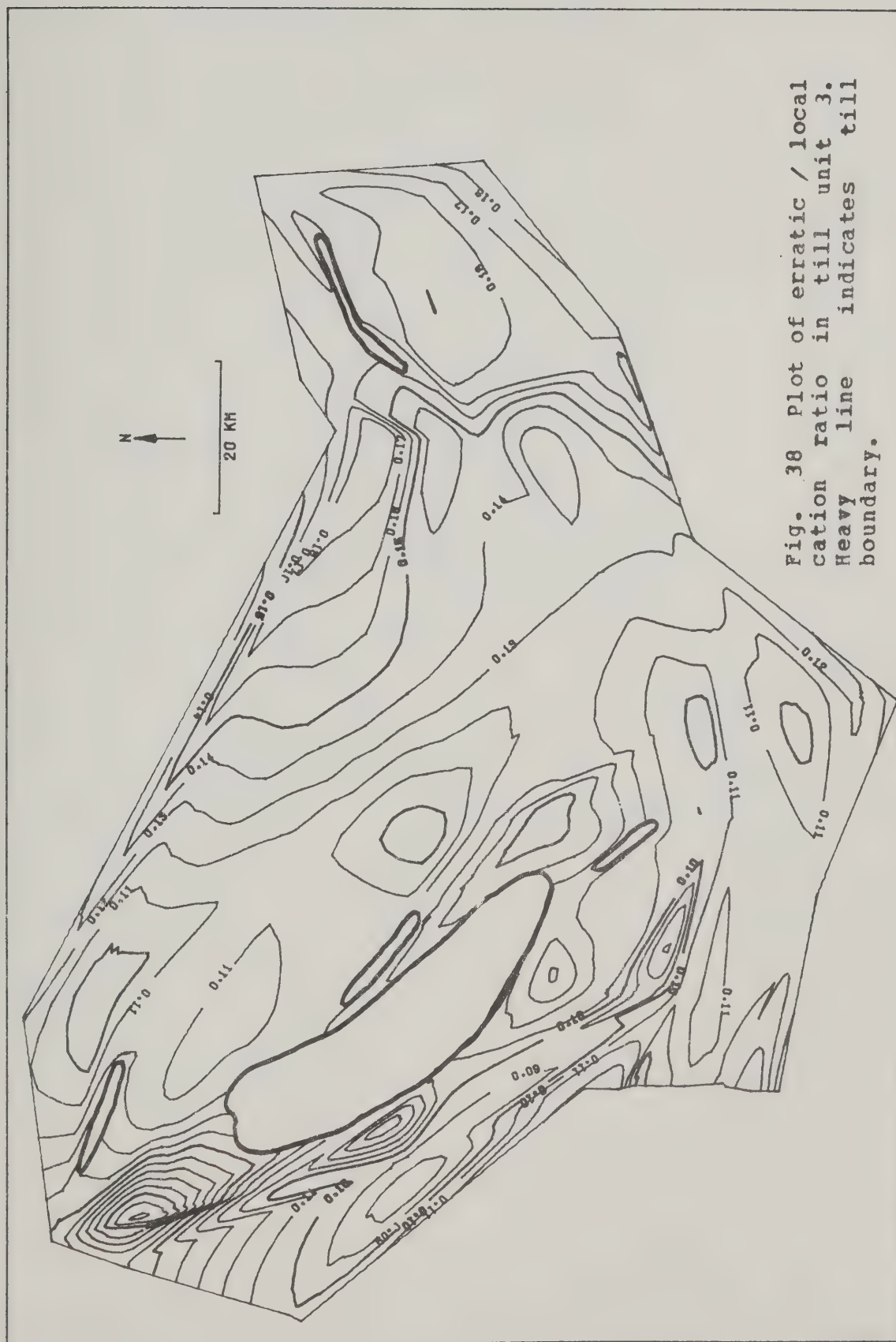
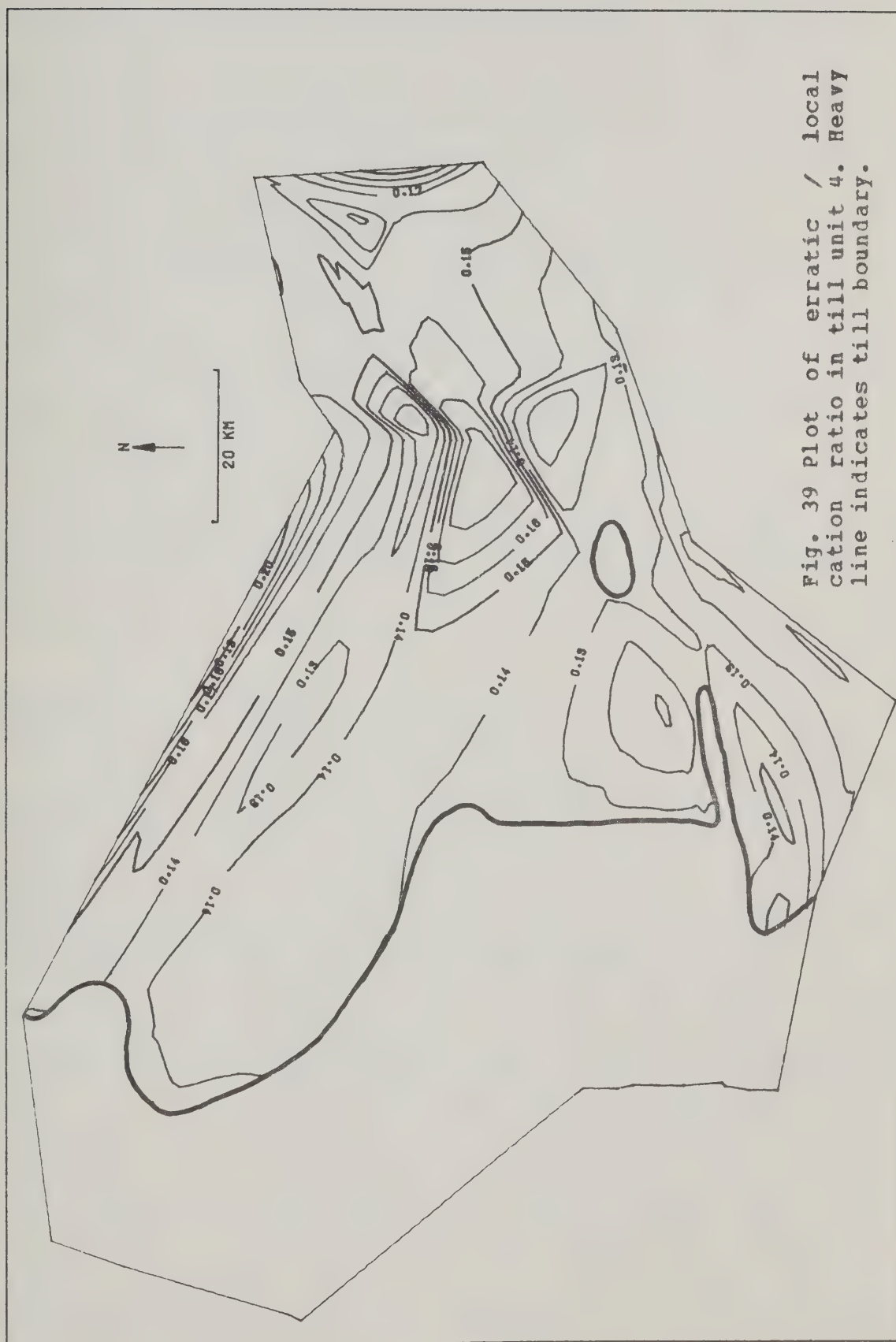


Fig. 38 Plot of erratic / local cation ratio in till unit 3. Heavy line indicates till boundary.









variation in carbonate content as a function of grain size and distance of transportation from the source. They found that with increased distance the original relatively coarse material was subject to more crushing, so that the maximum concentration of erratic material was found in successively finer size fractions until a stable "terminal grade" was reached. Thus conclusions based on one size-fraction do not readily relate to those based on a different fraction. Since both calcium carbonate equivalent and bulk chemistry are measured on the silt plus clay fraction in this study the regional variation is clearly due to dilution by local material in the down-ice direction rather than by further grinding action. This holds true for all till sheets.

#### 7.10 Local Compositional Variation

Due to poor sample distribution available for the analysis of variance of Appendix 9 it was not possible to compare the middle till, with its distinctive high erratic components against the other units. The two more extensive tills that were compared were fairly similar in their composition, since only the 40 to 60 mesh sand and the clay content differed significantly at the 5% level between all units over the map area. The sand figure is difficult to interpret, but the clay figure reinforces the visual observation that clay values often changed appreciably between units. Significant variation between the top and bottom samples of a till was only found for Na at the 5%



level, and for 20 to 40 mesh sand, silt and Mg at the 10% level. Considering the small number of samples and the low number of significantly varying analyses, it seems unlikely that a clear general change in composition of a till can be expected with depth. This agrees with visual impressions from the drill hole plots in Appendix 4 and the analysis of Appendix 8.

Various interaction effects were also found significant. The significant interaction of unit and sample position for 40 to 60 mesh sand, and carbonate grains, suggests that some units may be stratified with respect to these properties, but the lack of corroborative evidence makes this inconclusive, although the interaction between area, unit and depth found significant for sand and clay measurements suggests some textural stratification in some units in some areas. The interaction of area and unit is of interest, since it was found significant for all size analyses except 140 to 230 mesh sand, and clay. This strongly suggests that some areas of some units were markedly different in overall grain size from others, and that this effect, although affecting each till unit independently, was extensive enough to apply to a set of drill holes rather than just one. Examination of individual values shows that the upper till tends to be particularly coarse in the westernmost of the three areas examined, and the upper member of the lower till coarser in the easternmost area. Thus the texture of individual tills may





be fairly homogeneous not merely throughout their depth, but laterally for several miles. This interaction suggests that local depositional conditions responsible for depositing those till units that are compositionally fairly homogeneous may have an extent roughly comparable to the areas chosen for the analysis of variance design (10-20 km). Examination of the grain size maps in Appendix 7, as well as the analysis of variance results, suggests that these medium-scale variations in composition may be associated with the topographic features encountered by the moving glacial ice.

Gillberg (1965, 1967a,b) did extensive work on the relationship of the erratic component of till to the underlying topography. He clearly showed an exponential decrease of erratic material with increasing distance from the source material, primarily due to dilution with local material. He also demonstrated the appreciable variation in local ice direction in the lower part of an ice sheet (and hence in local till composition) in response to sub-ice topography, even though the upper, major portion of the ice sheet moved in the regional direction. He concluded that if the topographic slope was such as to impede the ice flow the erratic material would be deposited rapidly, whereas with a down-ice slope, erratic material would be transported greater distances. This corresponded to up-ice dips producing zones of deposition of both the local and the erratic material being transported, and down-ice dips producing zones of erosion of local material. In the lower



member of the lower till in the study area for this work the erratic component tends to be lower in the Helina and Beverly Channels and higher on the St. Lina slope, suggesting that, according to Gillberg, the former would be zones of easy flow and the latter a zone of resistance. The Kikino channel did not exist at that time. In the upper member of the lower till the erratics tend to be low in parts of the Beverly and Kikino Channels, suggesting fairly free ice flow in these, but much of the Helina channel shows little influence on the till composition, as much of it had previously been filled in.

The middle till varies widely in its erratic content and displays little pattern, and the upper till, although decreasing fairly systematically in erratic content towards the southwest, tends to have low values on the high ground and higher values in the residual low areas. No obvious explanation is available for this, although the erratic behaviour of the middle till may be due to its termination in the map area and consequently varying modes of deposition could well have been in effect at various times and places.

Thus, although medium-scale lateral variation in till composition may well reflect local depositional conditions controlled by the effect of the underlying topography on the continental ice sheet, more detailed sampling would be necessary to confirm this.



## 8.0 PREGLACIAL GEOLOGICAL HISTORY

### 8.1 Cretaceous Stratigraphy

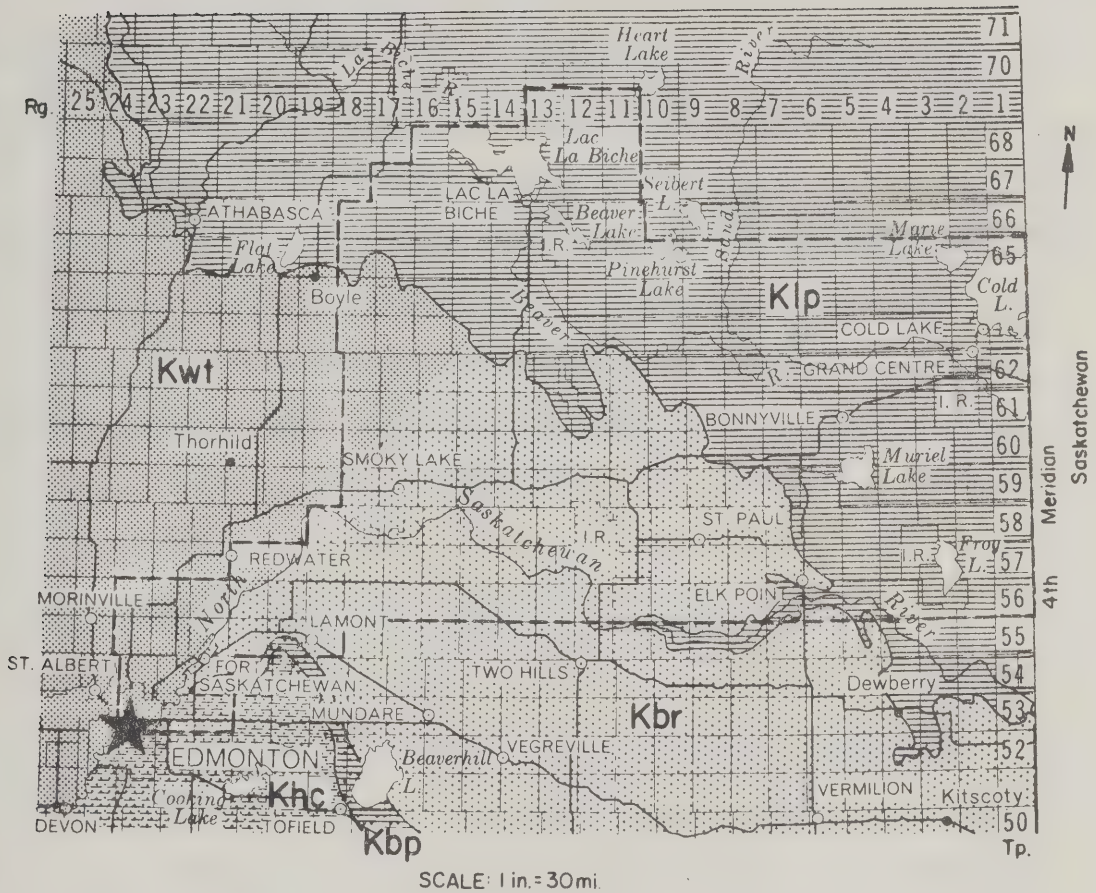
Two Upper Cretaceous bedrock formations directly underlie the surficial deposits in the map area (Carrigy, 1972). These are the Lea Park and Belly River Formations (Fig. 40). They dip gently to the southwest, with the younger Belly River only present in the southwest quarter of the map area. The Lea Park Formation consists of massive, marine, silty, dark grey shale and argillaceous siltstone with some fine-grained sandstone lenses. The Belly River Formation consists of non-marine, fine- to medium- grained, grey argillaceous bentonitic sandstone, grey clayey siltstone and grey or carbonaceous shale (Yoon and Vander Pluym, 1974).

### 8.2 Tertiary Stratigraphy

Westgate (1968) cited McConnell (1886) as recognizing the presence of elevated, commonly gravel-capped plateaux in the Plains of Western Canada. The implication was that the plateau surfaces once formed the level at which rivers flowed, and uplift and subsequent degradation lowered the surrounding country some 600 m. Because drainage was to the east, material clearly of Cordilleran origin was deposited throughout the plains. These formed the Saskatchewan Sands and Gravels which are well described by Stalker (1968). Although their lithology indicates this origin, rock and







### LEGEND

#### Upper Cretaceous



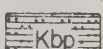
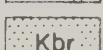
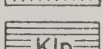
 Kwt	Wapiti Fm.	 Khc	Horseshoe Canyon Fm.
		 Kbp	Bearpaw Fm.
		 Kbr	Belly River Fm.
		 Klp	Lea Park Fm.
		---	Boundary of Study Area
		—	Geological Boundary

Fig. 40 (From Yoon and Vander Pluym, 1974) Bedrock geology of the Edmonton - Lac La Biche - Cold Lake area.





mineral species resistant to weathering and erosion predominate. The immediate local sources of these gravels in the Edmonton area were the late Tertiary conglomerates that covered former extensive preglacial uplands (Rutherford, 1937). Much of this material was deposited in the preglacial valleys as the uplands were denuded. Stalker (1968) provided criteria for their recognition, including the requirement that they lie between the bedrock and the first glacial deposits. These gravels in the buried valleys form a major groundwater source in the map area.

### 8.3 Bedrock Channels and Preglacial Drainage

Three major buried channels have been located in the test area by previous workers - the Beverly, Helina and Kikino Channels. (Yoon and Vander Pluym, 1974). The Helina Channel is extremely broad, with low slopes to the valley walls, its shape suggesting a mature major river valley. The Beverly Channel is less broad and has steeper bedrock walls (see Fig. 25). Preglacial material has been identified in the laboratory from the Helina Channel (drill hole 833E) and the Beverly Channel (706E and 831E). Field identification of preglacial gravels without laboratory confirmation occurred in holes 808E and 809E in the Helina Channel. The Kikino Channel is fairly narrow, steep-walled and cuts across the neck between the other two channels. No preglacial material has been found in the Kikino Channel, although a field identification in hole 730E was rejected in the laboratory



as being contaminated with Shield material.

To the east the merged Helina and Beverly Channels pass through Cold Lake and join a known bedrock channel that drains east (Christiansen and Whittaker, 1974). The known Vermillion Channel which enters the southeast corner of the map area also joins this river system in Saskatchewan. The smaller Vegreville Channel joins the Beverly just east of the Kikino Channel. The Beverly Channel is followed by the current North Saskatchewan upstream from the map area.

The Beverly Channel presumably contained the preglacial North Saskatchewan River (Westgate 1969, p.133). The Helina Channel probably contained the preglacial Athabasca River and possibly even the headwaters of the Peace River which may at one stage have drained east through Lesser Slave Lake, Calling Lake and Lac La Biche. (see Fig. 1 and McCrossan and Glaister, 1966) The Kikino Channel is thought to be of an early ice-marginal origin, draining water from the ice front and the North Saskatchewan northwards through Lac La Biche. The region around Lac La Biche possesses some very thick (greater than 100m) surficial deposits of gravel that is either of glacial or reworked preglacial origin. This valley deepening and gravel accumulation may be due to the Kikino ice marginal channel intersecting the preglacial Athabasca or Athabasca / Peace river system. Further drilling would have to be done to the north and west to resolve this problem. Surficial topography suggests that the preglacial Athabasca may have entered the map area from the



WNW, via Fawcett Lake and Calling Lake. The Kikino ice marginal channel may have continued NNW via Wandering River and Pelican Portage, associated with the breach in the mountain range now occupied by the Athabasca River flowing north towards Fort McMurray (see Fig. 1).

Recent work by Yoon et al. (1977) indicates the presence of an additional channel, the Primrose Channel, running just to the north of the study area between Imperial Mills (just to the north of drill hole 958E in Fig. 2) and the northern edge of Cold Lake (near hole 765E). Preglacial sands and gravels have apparently been identified in it. The preglacial Athabasca River may therefore have flowed east from near Lac La Biche to Cold Lake by either the Primrose or the Helina Channel at any given time. Further work is needed before confident descriptions may be made of the preglacial drainage in the area.





## 9.0 GLACIAL HISTORY

### 9.1 Lower Member of the Lower Till and Kikino Channel Formation

The sequence of maps of Figs. 25, 27 and 41 to 59 summarizes the glacial history of the area. Fig. 25 shows the current bedrock surface drawn by hand, utilizing not only information derived from the 78 drill-holes in Appendix 4 but data from other drill-holes and bedrock maps from other sources (Carlson 1967, Yoon and Vander Pluym, 1974). Fig. 27 is a computer drawn map of the same surface produced from the 78 drill-holes alone, as are the rest of the maps. The Helina and Beverly channels drained east through Cold Lake prior to glaciation, but the Kikino Channel in the west had not yet been cut. Fig. 41 shows the current thickness of materials deposited underneath the first glacial till whose bottom contact is shown in Fig. 42 and thickness in Fig. 43. This material includes preglacial sands and gravels in the Helina Channel as well as material deposited during the glacial advance. It can be seen from Fig. 43 that the Helina channel was partially filled with till at this time, and the Beverly channel may have been treated similarly. The ice depositing this lower member of the lower till seems to have reached most of the way up the St. Lina slope (at the junction of the Beverly and Helina Channels) and stopped. Since this would have blocked the drainage of the preglacial North Saskatchewan it was forced to cut the narrow ice-



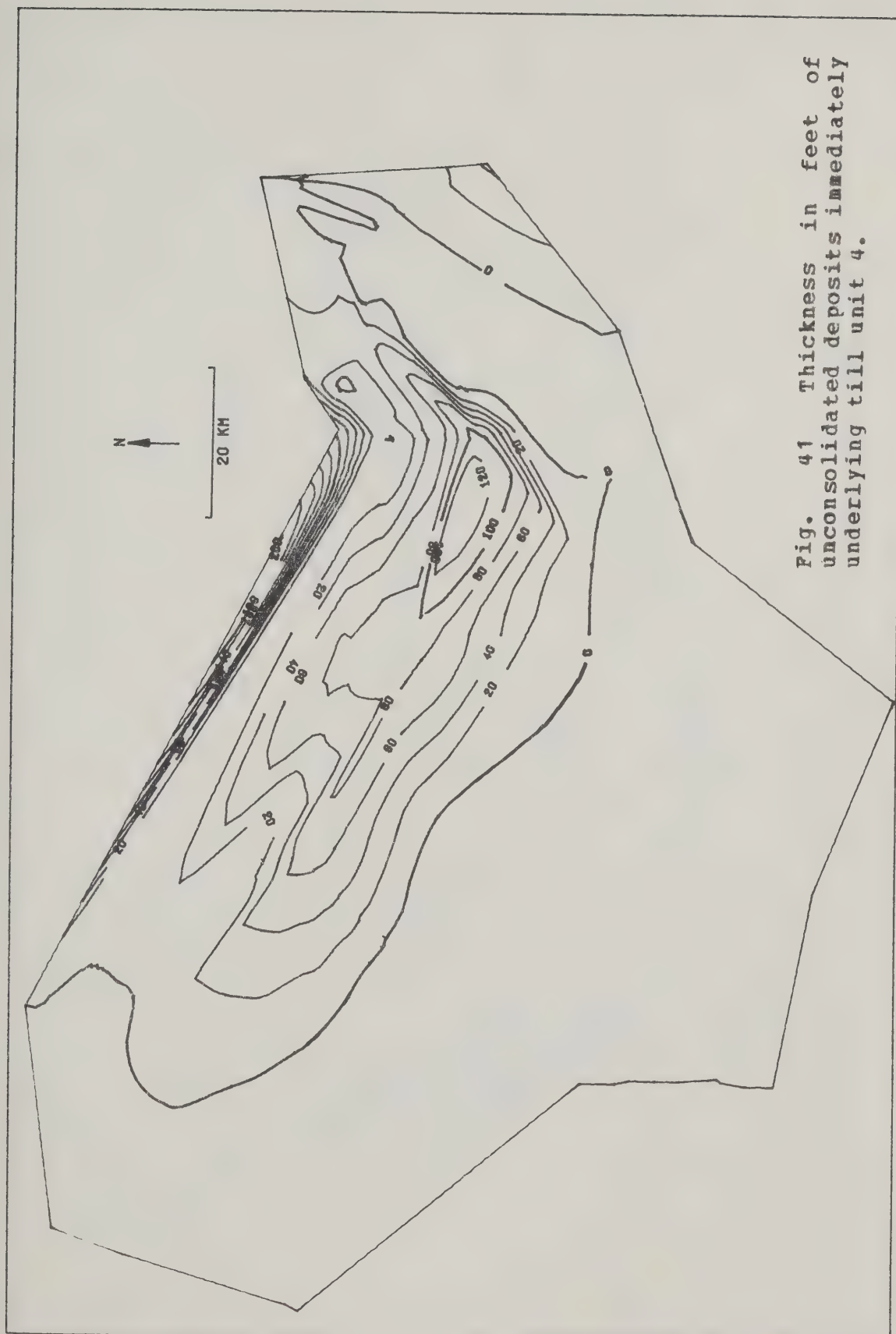


Fig. 41 Thickness in feet of unconsolidated deposits immediately underlying till unit 4.



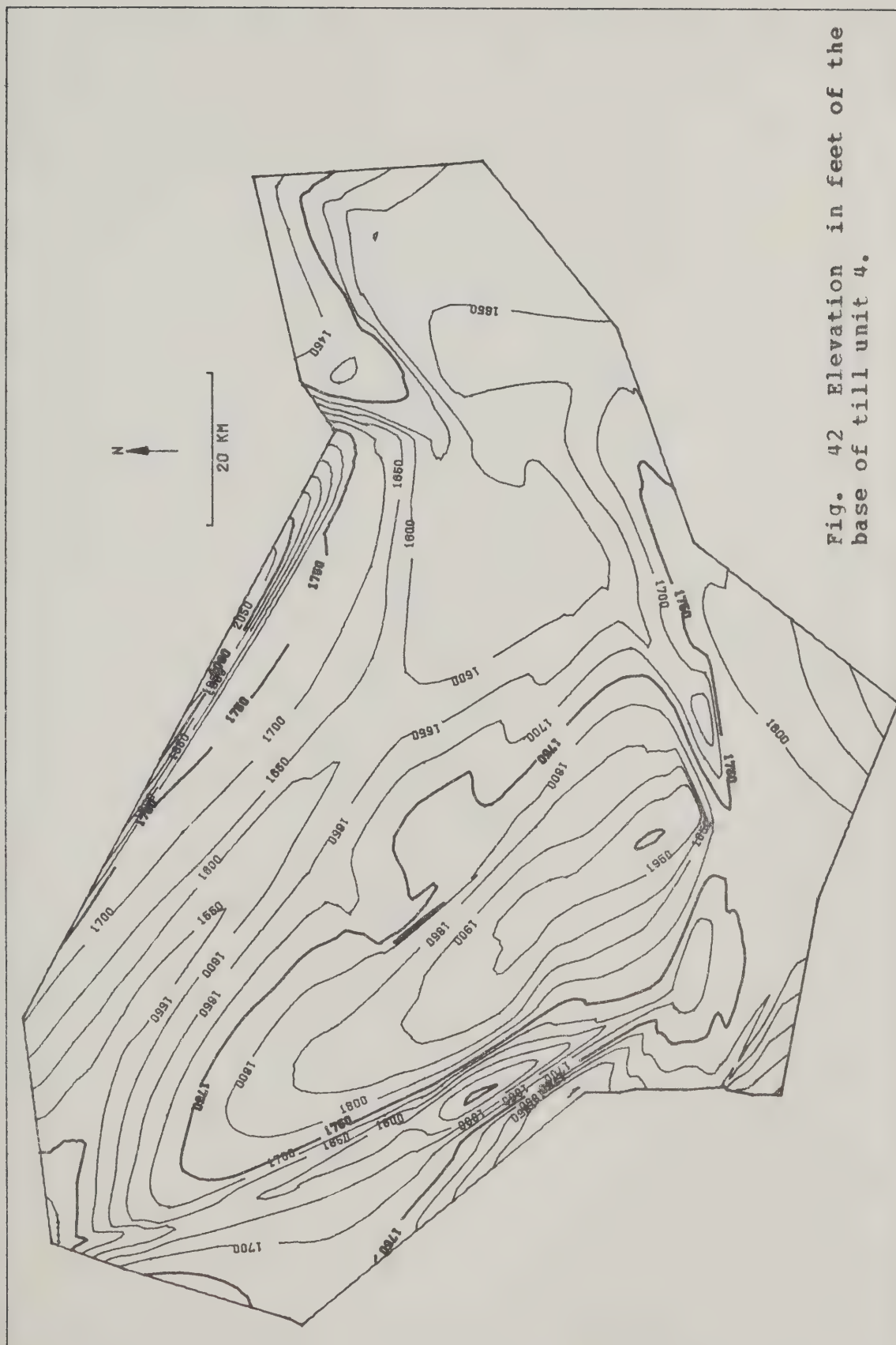


Fig. 42 Elevation in feet of the base of till unit 4.



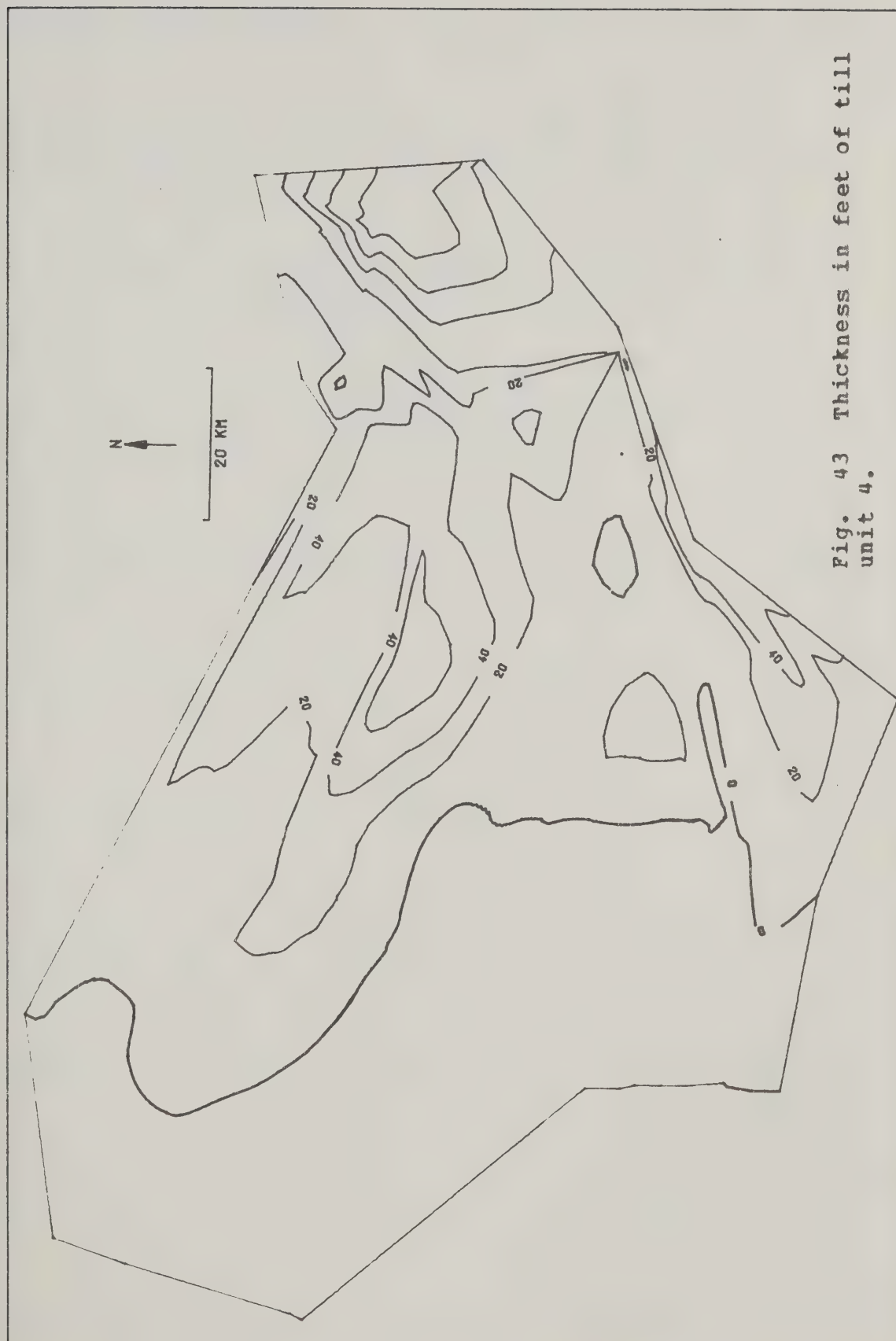


Fig. 43 Thickness in feet of till unit 4.





marginal Kikino Channel in order to drain northwest through Lac La Biche. The preglacial Athabasca river would have been treated similarly and the water of both dammed rivers, together with the ice run-off would have been channeled northwards towards Fort McMurray as described earlier.

Although it is usually considered that ice-marginal channels are supplied predominantly by ice run-off, the Kikino channel does not appear to be present to the south of the Beverly channel. Although ice-shoved bedrock is common in the area, the bedrock topography map (Yoon and Vander Pluym 1974, Fig. 6) does not indicate even a blocked route for a southern extension of the Kikino channel, strongly suggesting that this channel is in fact due to redirected river drainage. This idea was also made by Rutherford (1941) to explain the general drainage pattern of the Saskatchewan River system, where most channels show an abrupt bend to the southeast suggesting that drainage, being blocked by ice, was forced to flow along or roughly parallel to the ice front. Ultimately the channels turn again to the northeast, indicating that the regional slope was still in that direction. In the same paper the author showed the northwestern extension of the "Altamont" moraine associated with the Coteau in Saskatchewan passing along the more recently discovered Kikino channel. Another morainal system, the "Viking", is shown passing in a north-south direction a few miles further to the west. This suggests that one, if not two, regional glacial advances stopped near the western



portion of the map area.

### 9.2 Deposition of the Upper Member of the Lower Till

Following the deposition of the lower member of the lower till the ice probably retreated to the northeast. On the basis of the estimated topography of the top contact of the lower member of the lower till (Fig. 44), the Helina channel seems to have remained blocked. The Athabasca continued to drain north, but it is uncertain whether the North Saskatchewan resumed its old course as suggested by Fig. 44, or continued to follow the Kikino Channel as suggested by the bottom contact of the upper member (Fig. 46). Fig. 45 shows the thickness of non-till material below the upper member and above the lower member in the eastern part, where the lower member ice did not advance. It is frequently impossible to tell, in the absence of a particular till in any one drill hole, whether the non-till material was deposited before, after or contemporaneously with the missing till. Consequently Fig. 45 and other maps of this type assume the material to be as young in age as the tills present will permit. In the case of Fig. 45 the deposits between the upper and lower members of the lower till therefore include in the west some preglacial gravels as well as some presumed glacio-fluvial material deposited in front of the ice.

Fig. 47 shows the thickness of upper member of the lower till deposited. It can be seen to cover the whole map



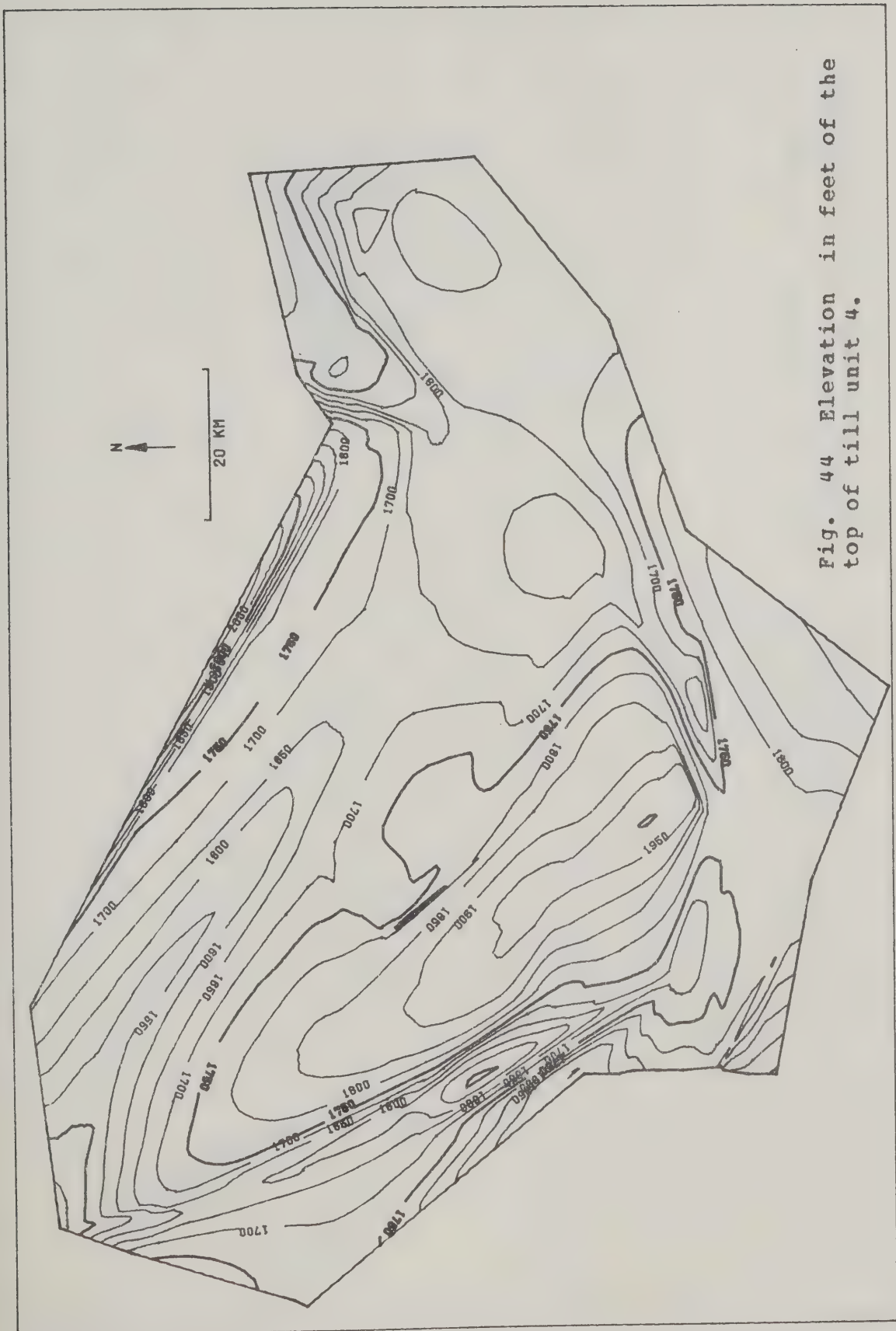


Fig. 44 Elevation in feet of the top of till unit 4.





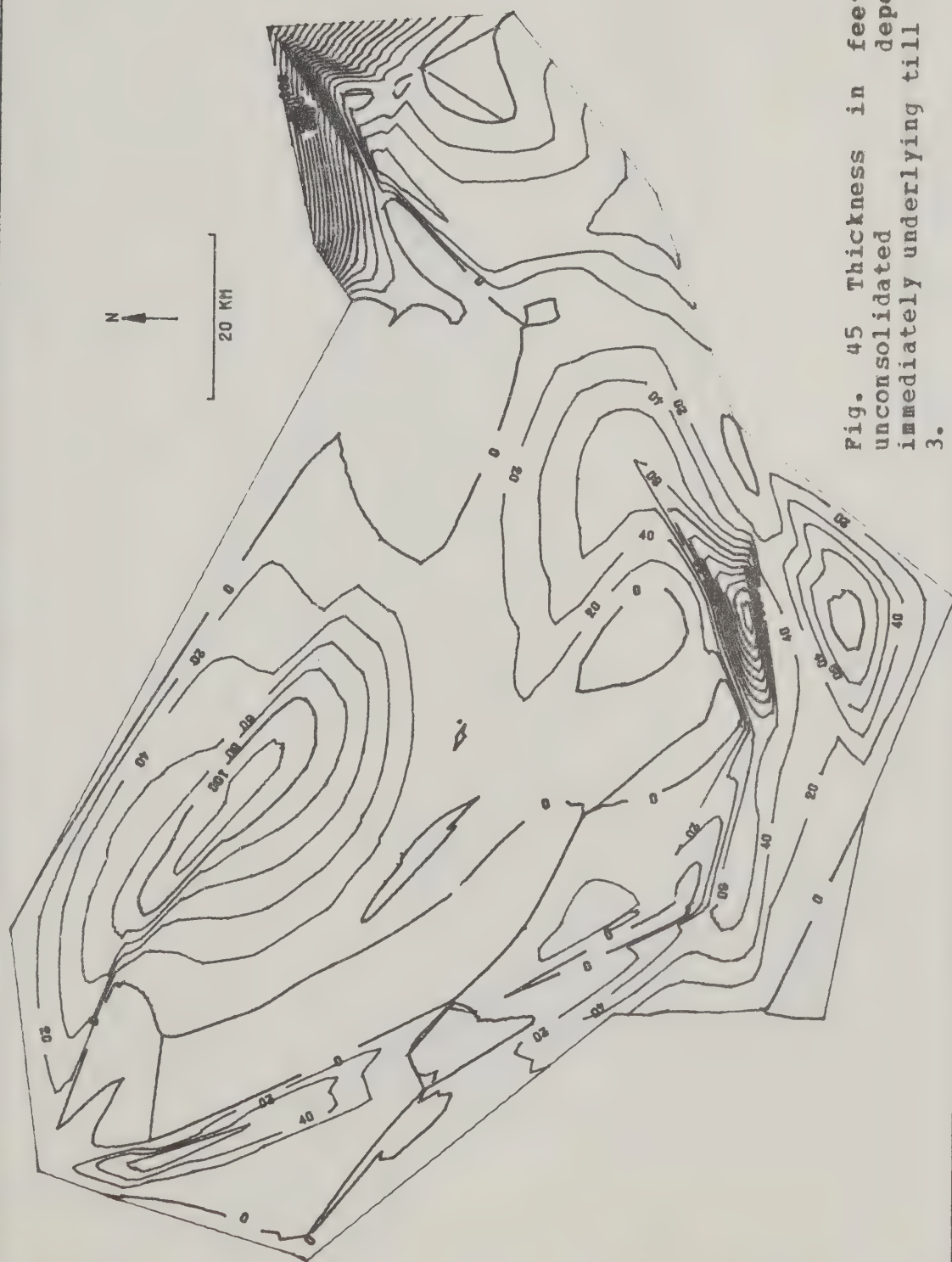


Fig. 45 Thickness in feet of unconsolidated deposits immediately underlying till unit 3.



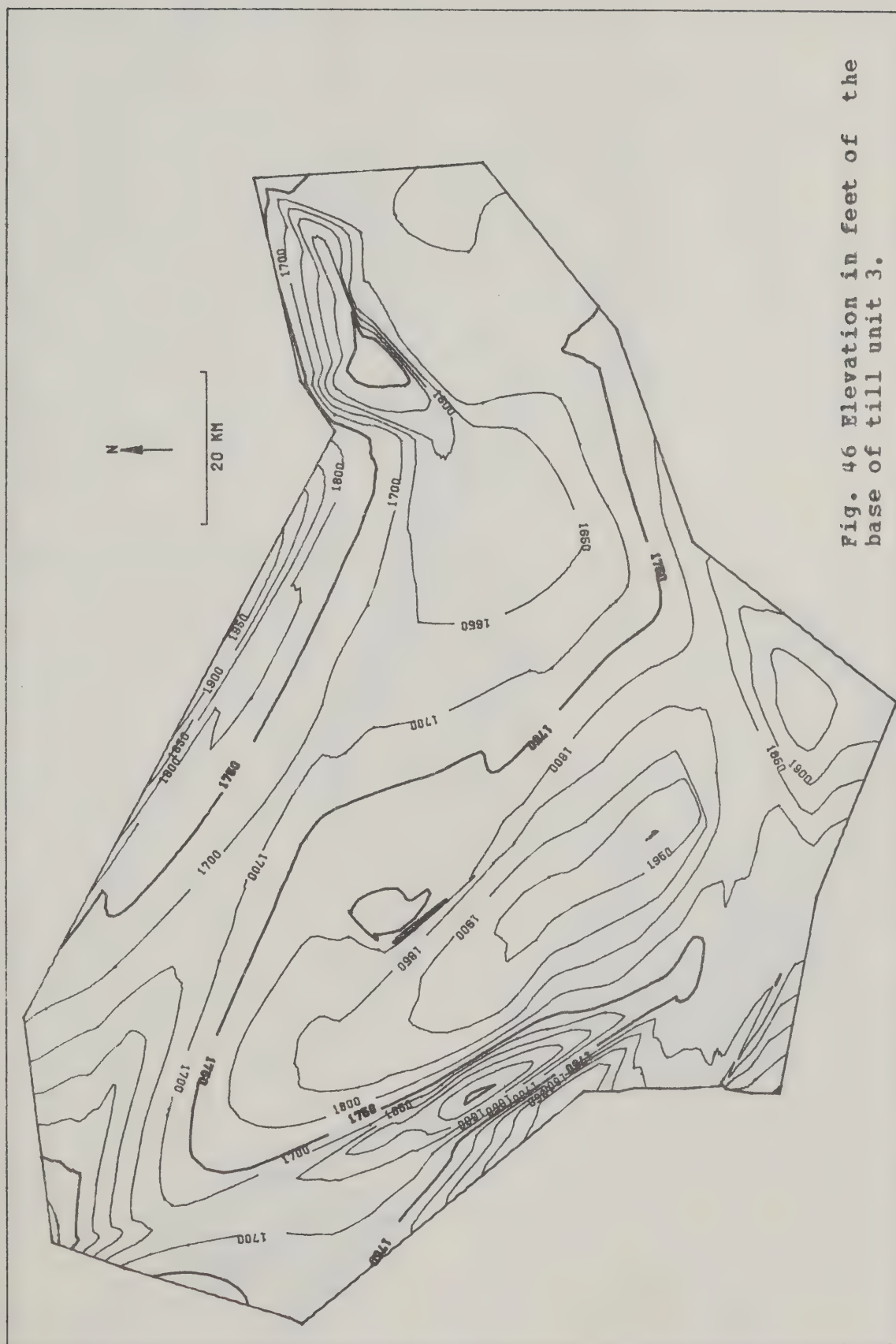


Fig. 46 Elevation in feet of the base of till unit 3.



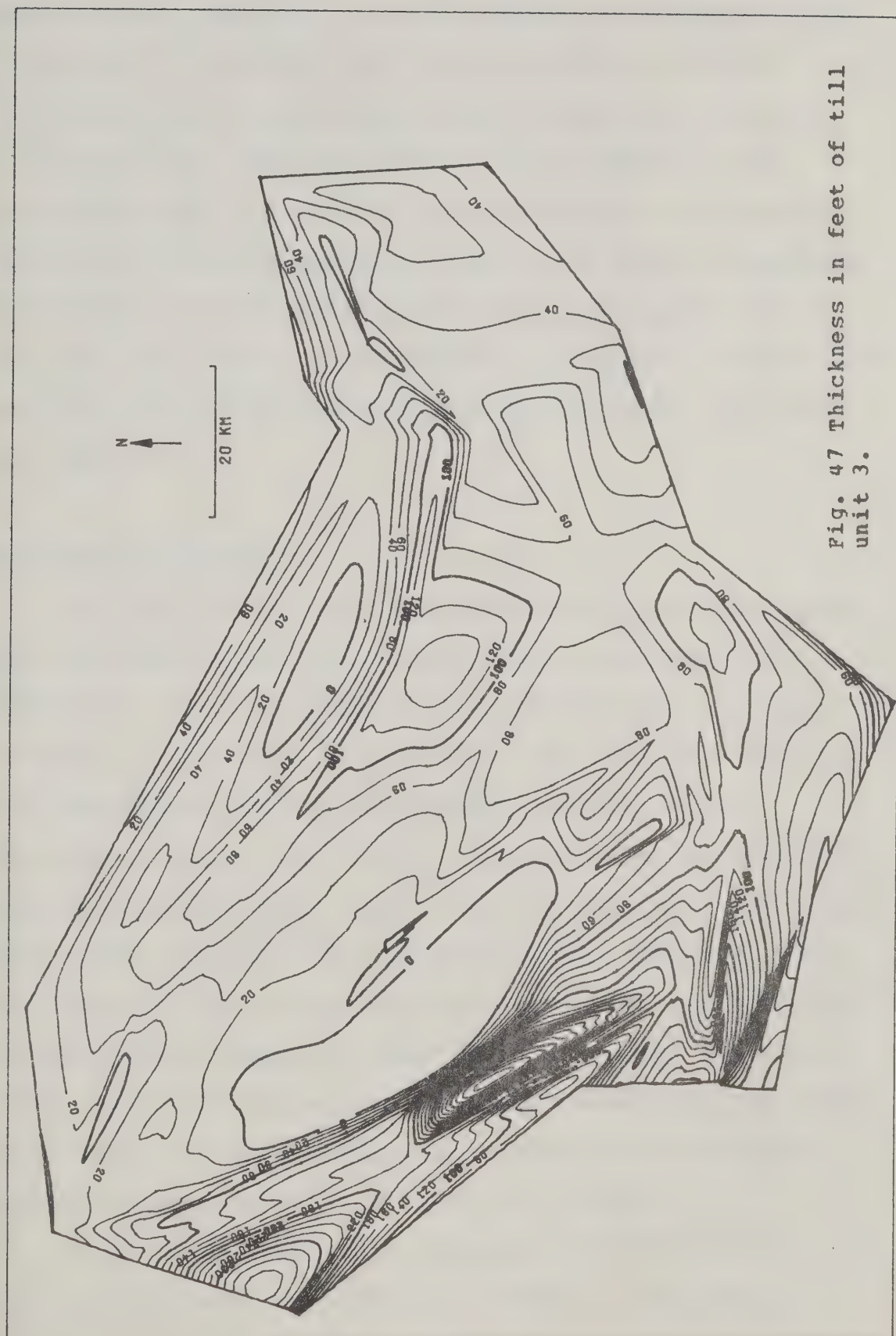


Fig. 47 Thickness in feet of till unit 3.





area with the exception of the Whitefish Lake bedrock high. Thicknesses of greater than 300 feet (100m) are found filling in the recently-formed Kikino Channel. This great till thickening suggests that the higher ground on the western boundary of the map area again formed an impediment to further ice movement. That this is the case is supported by the fact that some or much of the upper member till on the west side of the Kikino channel is composed of disturbed bedrock. This is the case for drill-holes 952E, 953E, 956E and 957E.

### 9.3 Bedrock Thrusts

St. Onge (1972) noted the hummocky moraine associated with the high ground just to the west of the region now associated with the Kikino channel. He remarked ( p. 93) "Ridges on both sides of Lake Amisk and south of Kikino are also associated with it. The ridges south of Kikino represent a large push moraine constructed by southwesterly flowing ice." While the current work suggest that the ridges were formed prior to the deposition of the uppermost till, both sets of ridges are associated with disturbed bedrock - the Lake Amisk ridges are near drill-holes 952E and 953E while the Kikino ridges are near drill-holes 956E and 957E. These morainal features may well be associated with the suggested Viking moraine of Rutherford (1941).

Of particular note are the sampled drill-holes 434E and 828E together with the other drill-holes in the cross-





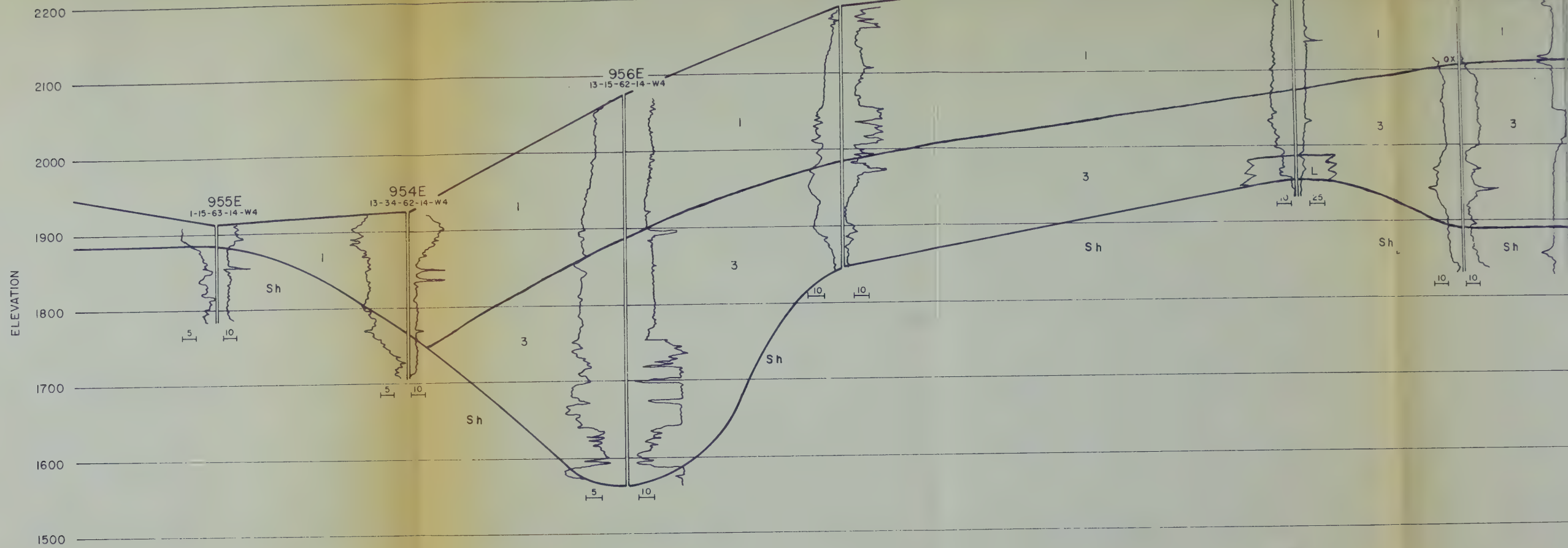
section of Fig. 48. On the basis of drill holes on either side of the cross-section (Yoon and Vander Pluym 1974, Fig. 6) the Beverly channel should intersect the cross-section near drill hole 434E with a valley floor below 1800ft (550m). However, the drill holes (which are numbered in chronological order) prior to number 828E had identified a bedrock plateau at 1850ft. between holes 435E and 436E, and a surprising bedrock high of approximately 2100ft. elevation where the channel was expected. The subsequent drilling of hole 828E to depths well below the first occurrence of bedrock material initially indicated that permeable material occurred below 130ft. of shale. This suggested that the missing valley floor had been found at this location, with a rather implausible bedrock high immediately to the north. Subsequent examination suggested that the bedrock high was in fact an ice-shoved bedrock slab of shale overlain by sandstone. The Lea Park Shale - Belly River Sandstone contact subcrops in this area (Fig. 40), and holes 759E and 796E to the south terminate in the sandstone while holes 735E, 754E and others to the north terminate in the shale. For this reason the first sandstone / shale sequence in holes 828E and 434E is interpreted as an upward-thrust ice-shoved bedrock slab. The second sandstone is interpreted as Belly River Sandstone in place, underlain by Lea Park Shale. Hole 795E also terminates in the Belly River Sandstone. It is thought that this bedrock slab was thrust across the Beverly Channel by the ice advance that deposited the upper



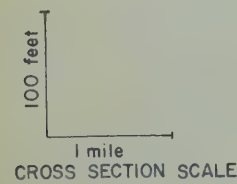
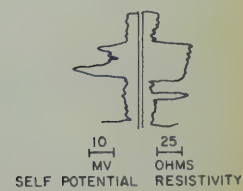
Tp 63  
Tp 62

Tp 62  
Tp 61

Tp 61  
Tp 60

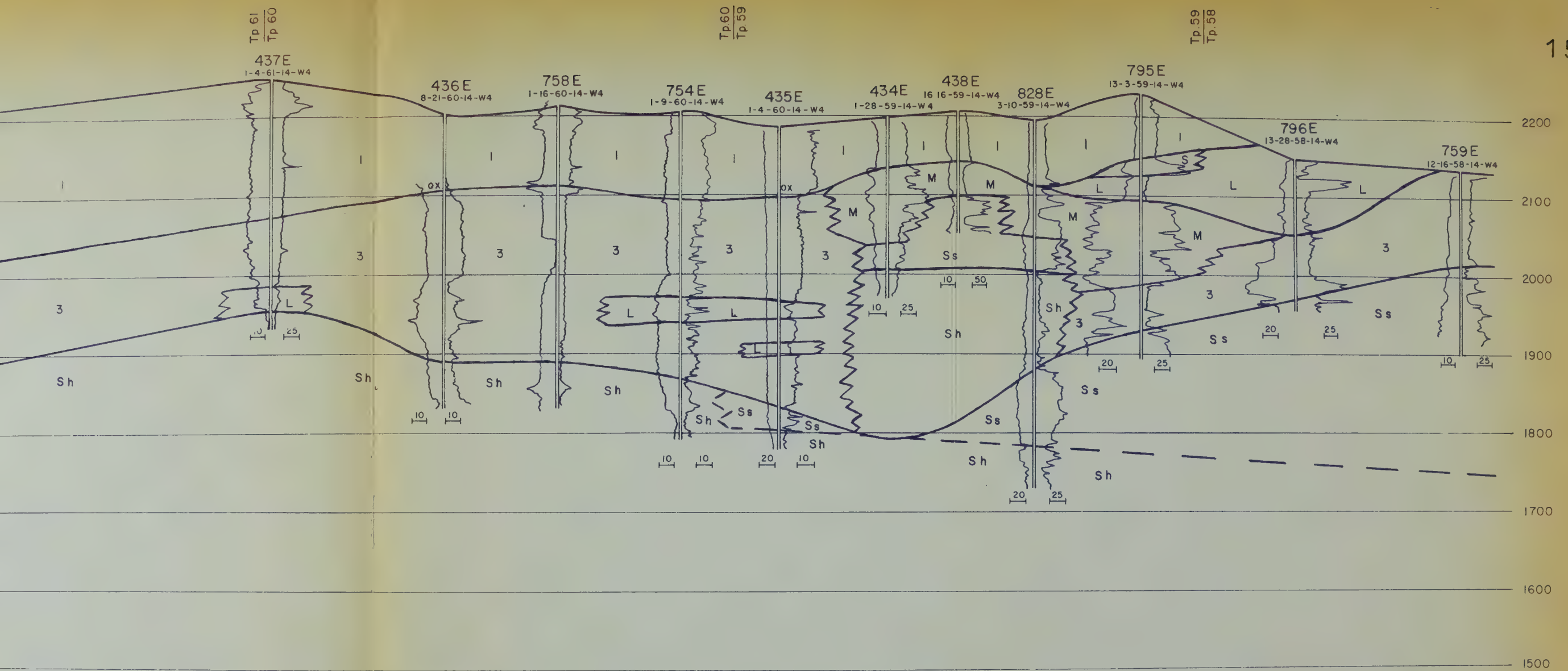


# ELECTROLOG SYMBOLS



# LEGEND

- 1 Till unit 1
- 3 Till unit 3
- S Sand &/or gravel



## LEGEND

I	Till unit 1	L	Lacustrine silt & clay
3	Till unit 3	M	Mixed sand, lacustrine, bedrock or till
S	Sand &/or gravel	Ss	Bedrock sandstone
		Sh	Bedrock shale

Fig. 48

North - south cross-section  
through the Vilna bedrock thrust.





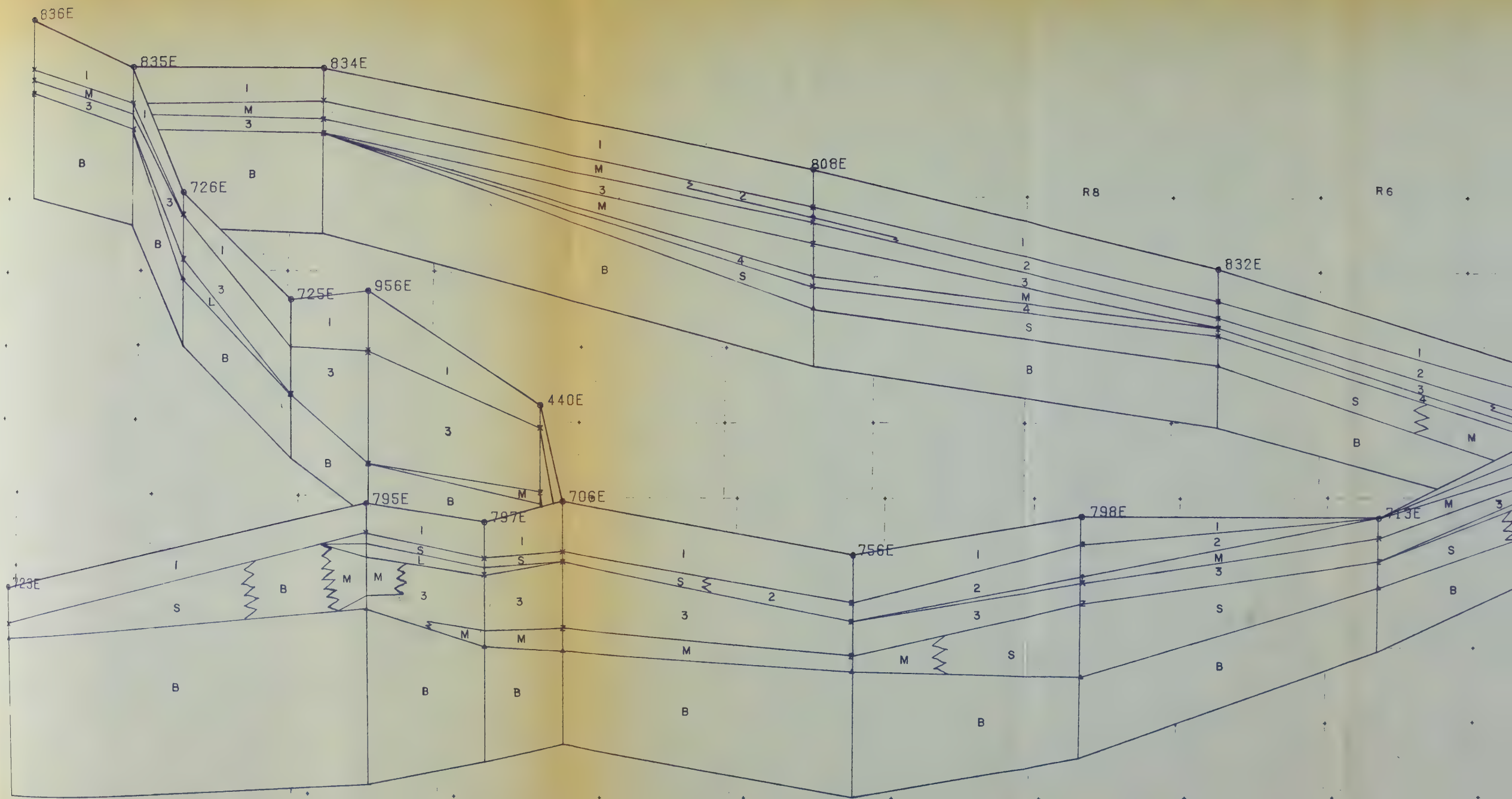
member of the lower till. This interpretation is supported by the presence of bedrock thrust ridges further north in the Kikino Channel (St. Onge, 1972) and also by extensive disturbed bedrock found in drill holes 952E, 953E, 956E and 957E.

The high ground formed by the bedrock thrust extends nearly to Wahstac and appears to have formed a permanent dam preventing the North Saskatchewan from following the Beverly Channel any further (Figs. 25 and 49). The river, thus dammed, was forced to cut an outlet to the south, eroding over 100 feet (30m) of bedrock in the process. It is not clear to what extent the upper member of the lower till extends to the west of the Vilna-Wahstao barrier.

While the separation of the lower till into upper and lower members may be uncertain on the basis of composition alone it is quite clear that some action, prior to the dominant upper member deposition and subsequent to the initial disruption of the preglacial drainage, must have caused the Kikino Channel to be formed, and a lower member ice advance as estimated from electric logs reached approximately the right location. An alternative, if similar, interpretation is that the upper member glaciation paused for an extended period of time just to the east of the current Kikino Channel position and the ice marginal channel was formed, only to be buried immediately the ice readvanced.

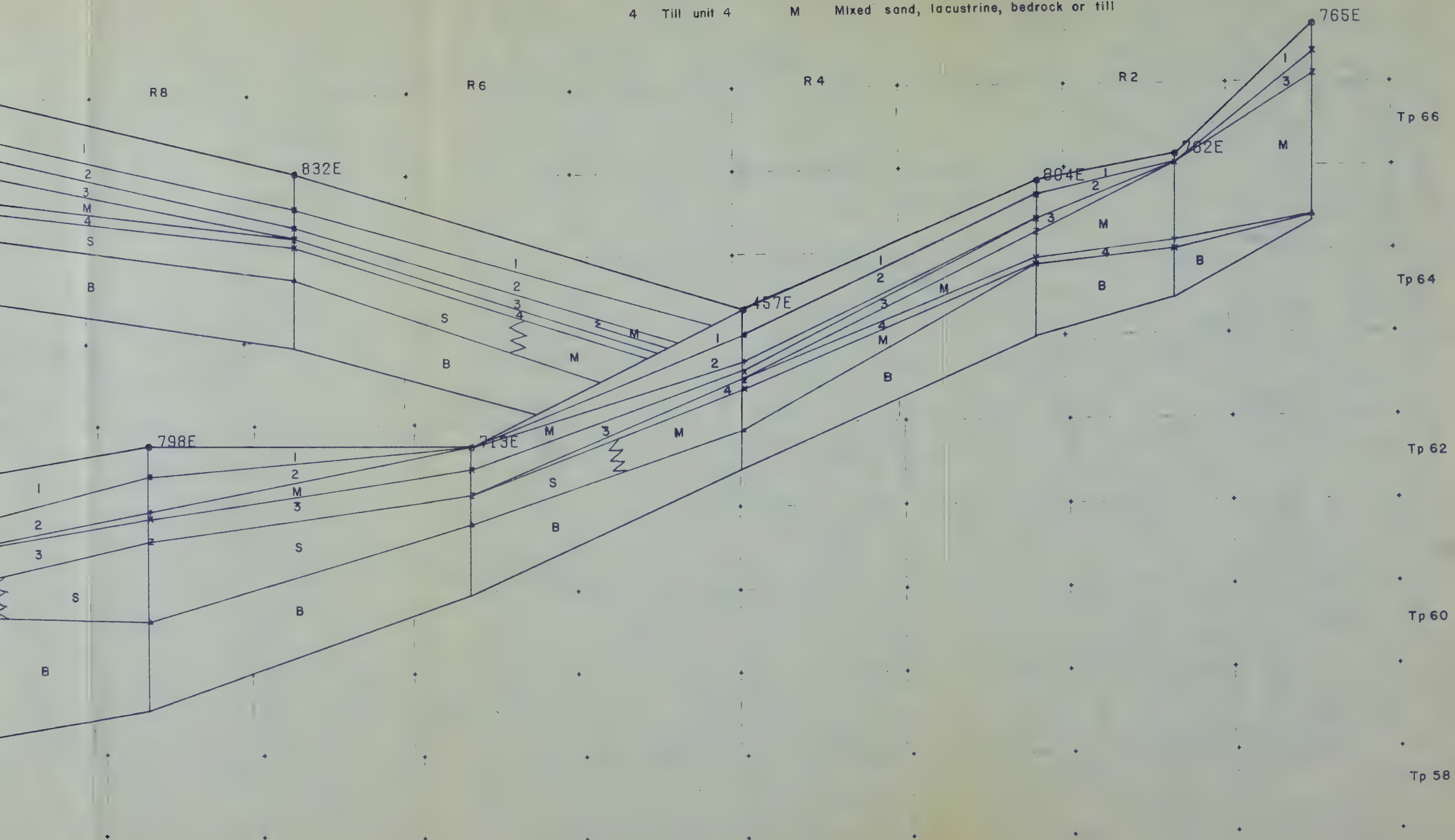






- |   |             |   |   |
|---|-------------|---|---|
| 1 | Till unit 1 | S | Sand &/or gravel                        |
| 2 | Till unit 2 | L | Lacustrine silt & clay                  |
| 3 | Till unit 3 | B | Bedrock sandstone or shale              |
| 4 | Till unit 4 | M | Mixed sand, lacustrine, bedrock or till |

Fence diagram along the  
bedrock channels of the  
study area.





#### 9.4 Middle Till Deposition

Fig. 50 shows the elevation of the top of the lower till. There is no evidence of major drainage in the area, although local streams may have flowed east to Cold Lake. Neither extensive leaching nor clear zones of oxidized iron associated with this till are found in the Sand River area. Fig. 51 shows the inter-till deposits directly below the middle till, consisting mainly of valley-fill deposits near Cold Lake, and Fig. 52 shows the pre-middle till surface. As seen from Fig. 53, the thickness of the middle till is greatest in the old Beverly Channel and at the top of the St. Lina slope, to the west of which it was not found in the drill-holes. This thickening, along with its topographic expression as shown in Fig. 54, suggests the formation of an end-moraine, possibly the Altamont of Rutherford (1941). Fig. 55 shows the inter-till deposits below the upper till - the four patches of thick deposits are clearly deposits whose true age is older but can not be so identified due to the absence of the required till layer at that location. Fig 56 shows the topography at the base of the upper till, and most of the weathered zones located are associated with this surface, indicating an extended ice-free period. There is no obvious drainage pattern.





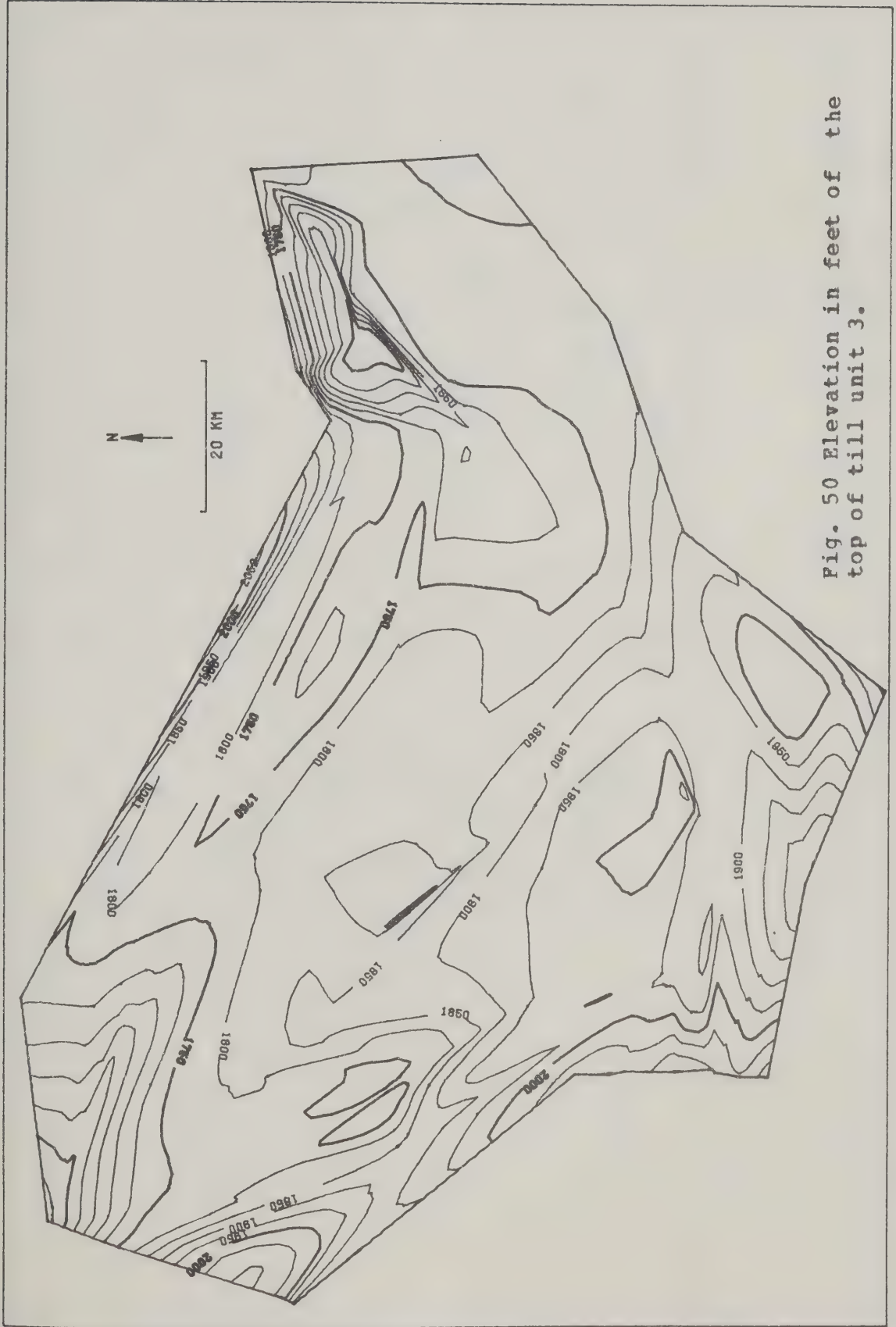


Fig. 50 Elevation in feet of the top of till unit 3.





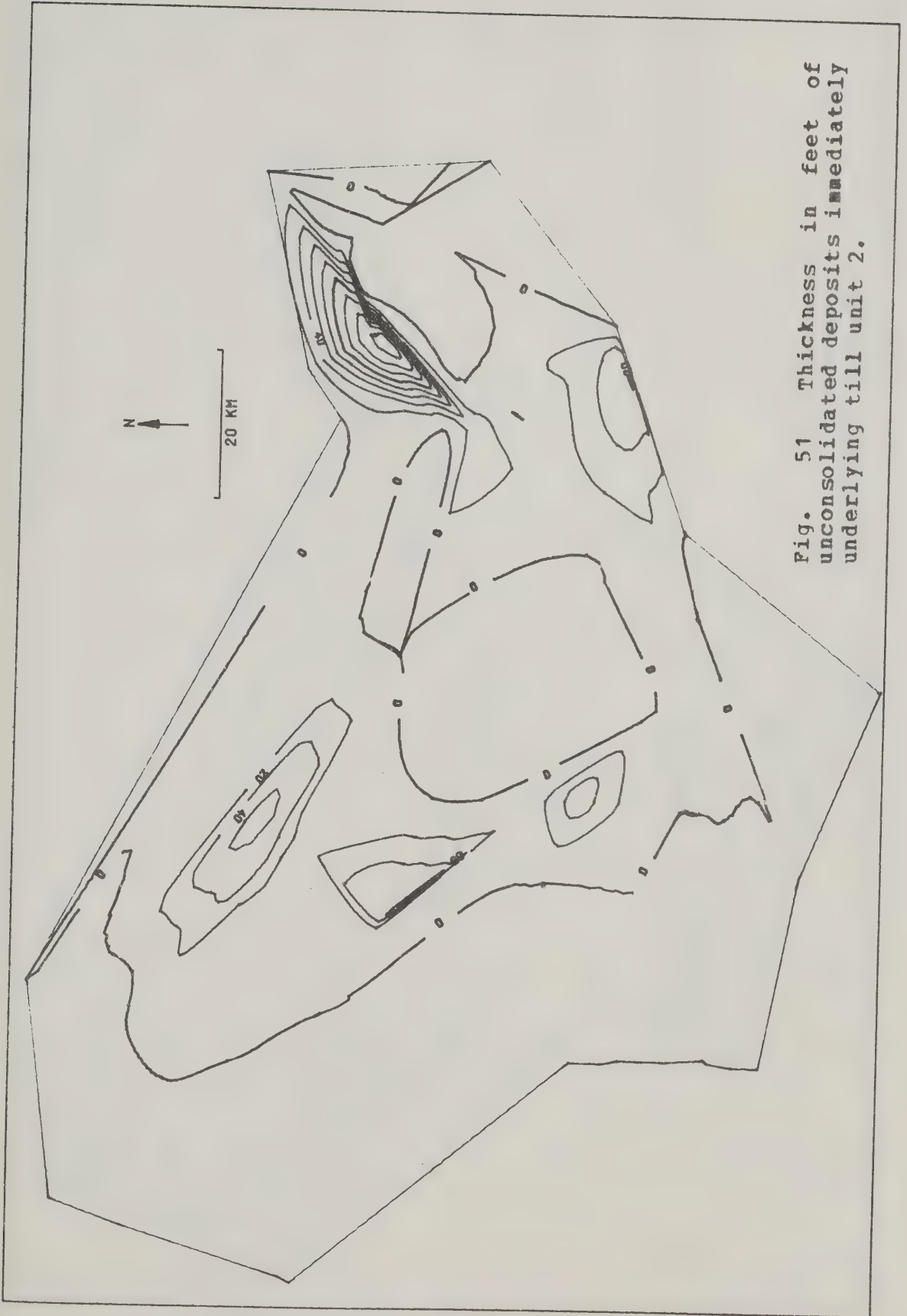


Fig. 51 Thickness in feet of unconsolidated deposits immediately underlying till unit 2.



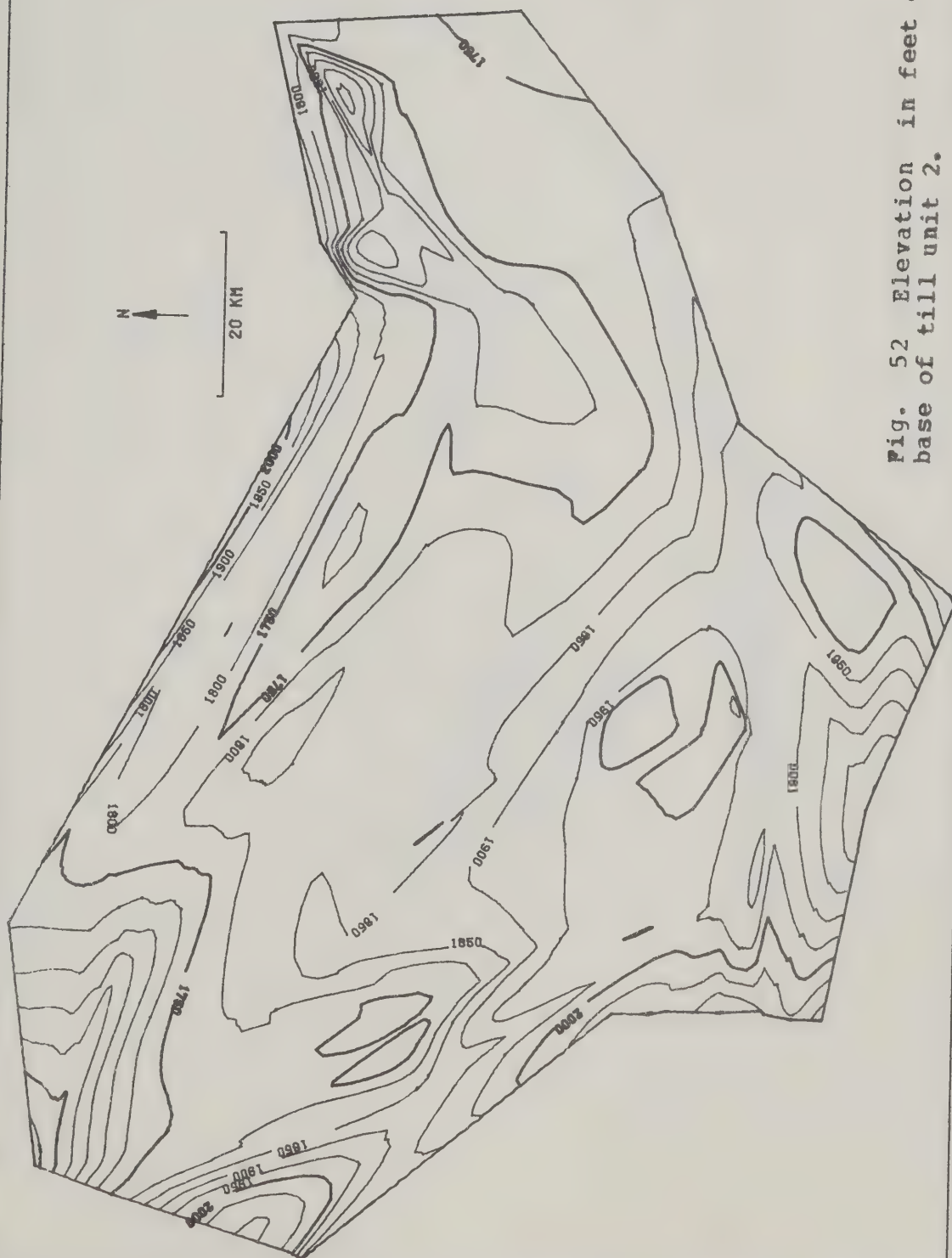


Fig. 52 Elevation in feet of the base of till unit 2.



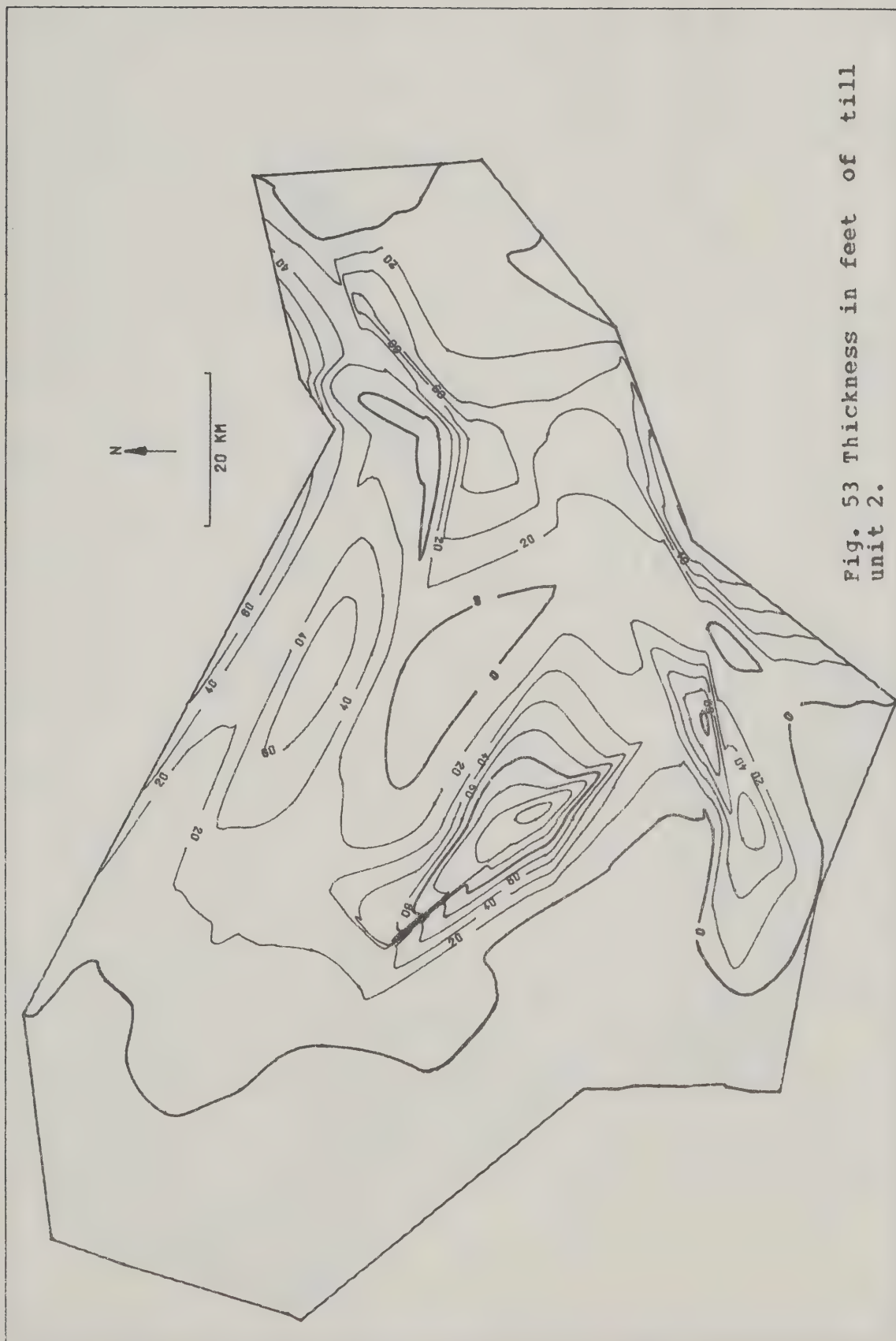


Fig. 53 Thickness in feet of till unit 2.





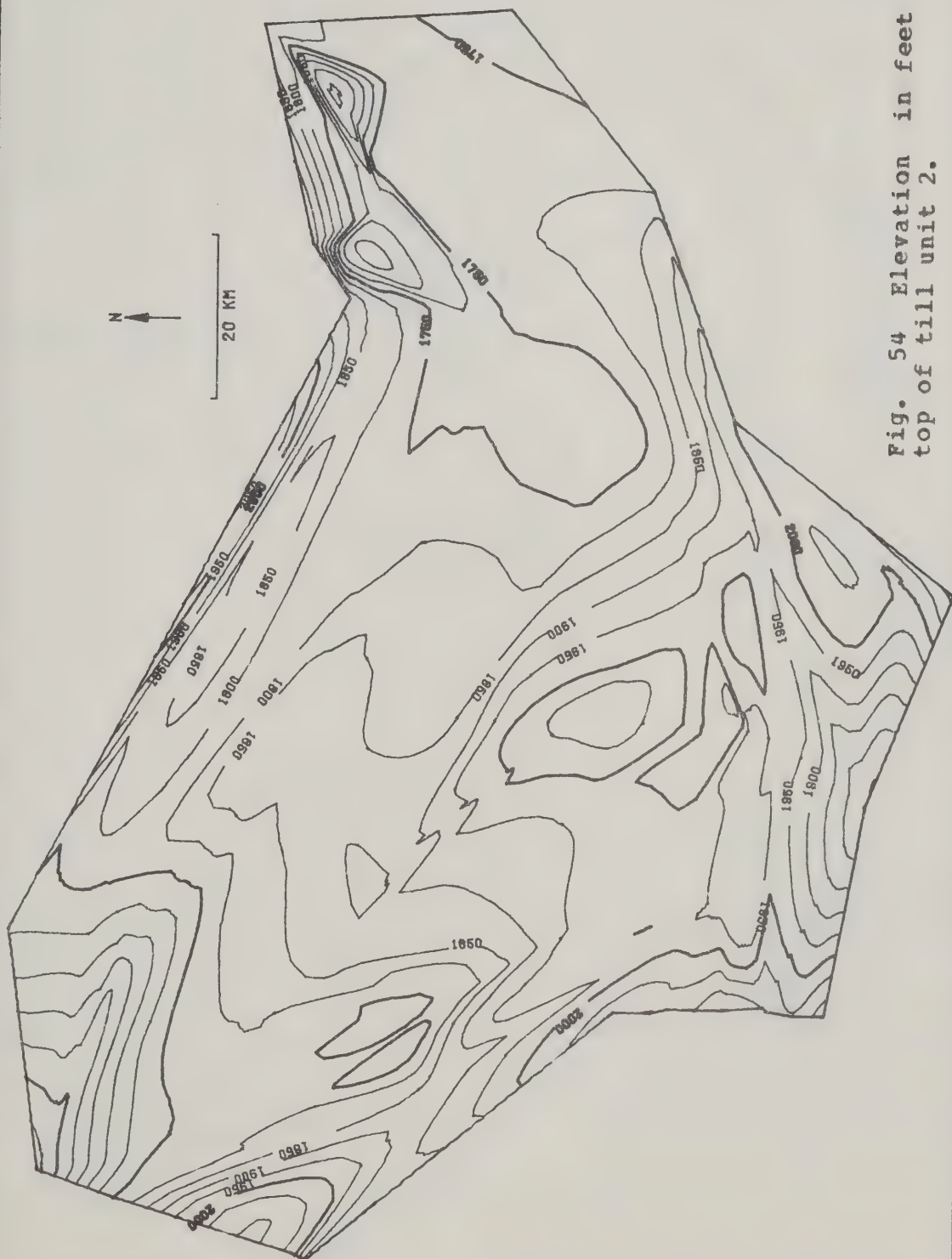


Fig. 54 Elevation in feet of the top of till unit 2.



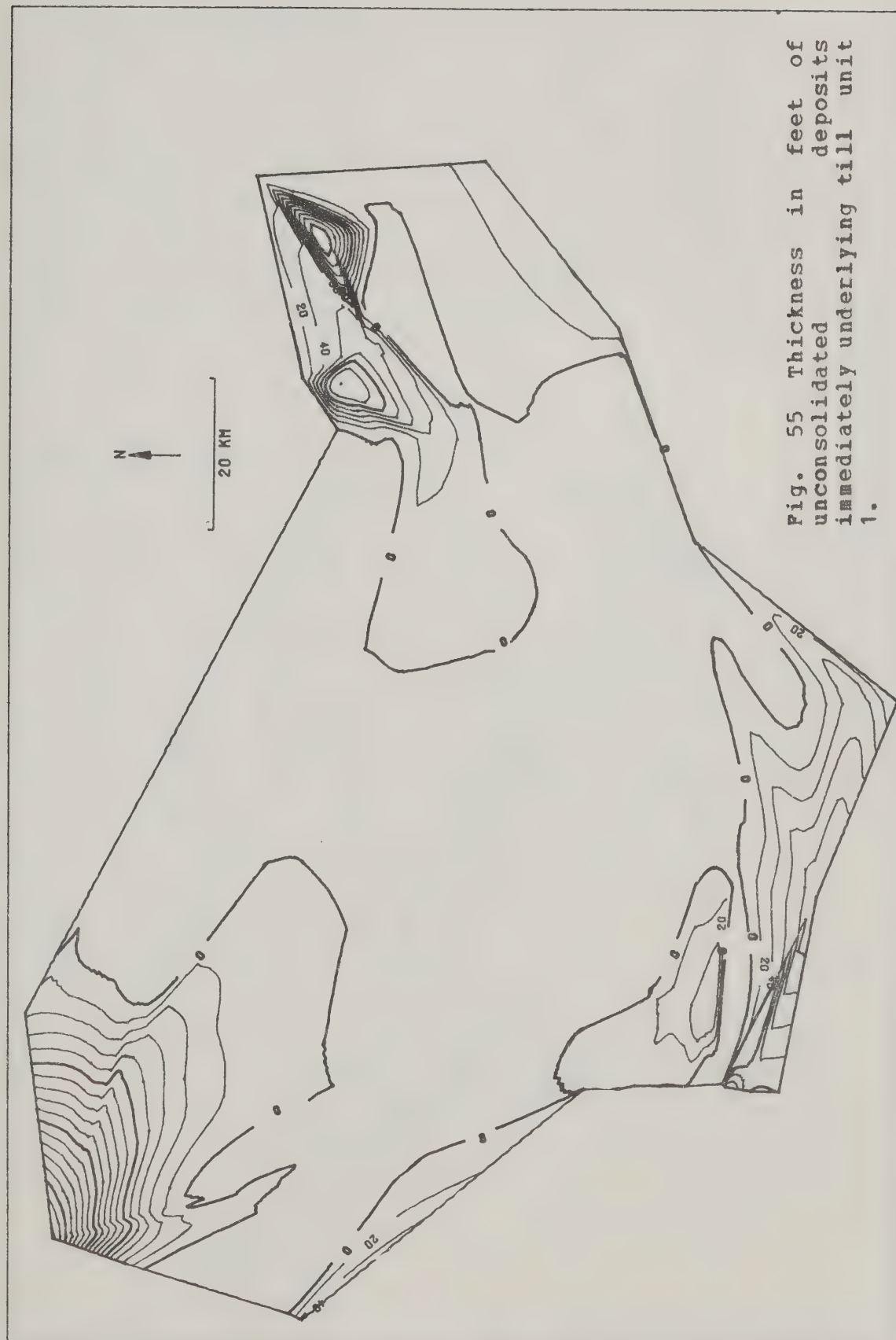


Fig. 55 Thickness in feet of unconsolidated deposits immediately underlying till unit 1.



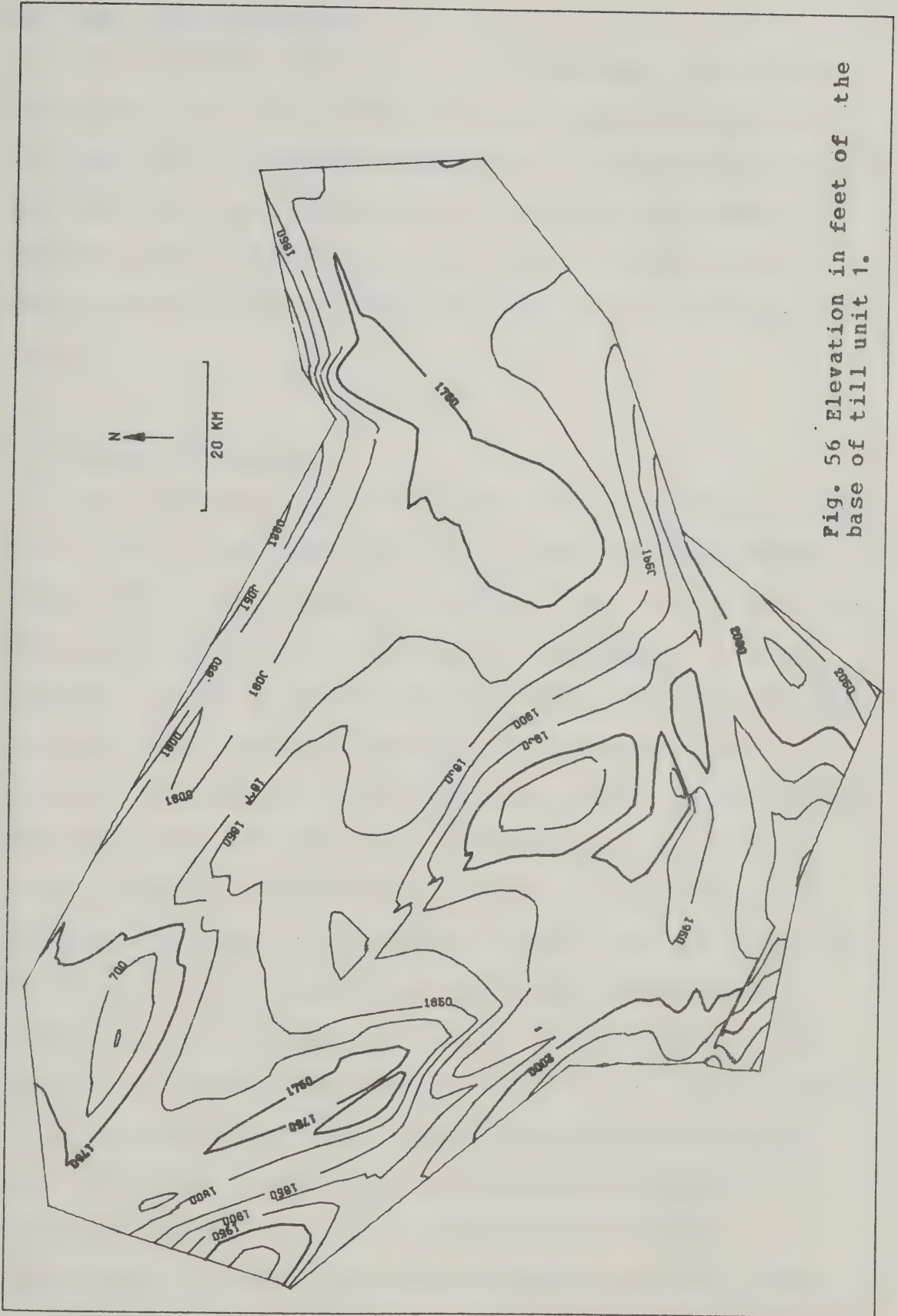


Fig. 56 Elevation in feet of the base of till unit 1.



### 9.5 Upper Till Deposition

Fig. 57 shows the thickness of the upper till, which extends over the whole area, and also continues further to the west. The predominant effect is of further deposition of till to reduce irregularities in the underlying terrain, together with a tendency for the till to thicken at the western edge of the map area where the ice met the higher ground.

### 9.6 Present Topography

Fig. 58 shows the elevation of the top surface of the upper till. It suggests that little major drainage passes through the area, although as shown on Fig. 59 and Fig. 2, (the generalized map of the present topography), a broad vestigial valley is still found following the Helina Channel between Lac La Biche and Cold Lake. The current rivers, primarily the Amisk, Beaver, and Sand Rivers, are relatively small with very localized drainage basins. A minor low region, occupied by Whitefish, Goodfish and Garver Lakes remains as evidence of the Kikino Channel and its effect on ice movement, but little evidence of the Beverly Channel remains. Indeed, a conspicuous neck of high ground still crosses the channel between Vilna and Wahstao, evidence of the magnitude of the Vilna bedrock thrust. Just upstream from this barrier, St. Onge (1972) noted a system of meltwater channels that may have drained the North Saskatchewan River to the north at one stage in the retreat





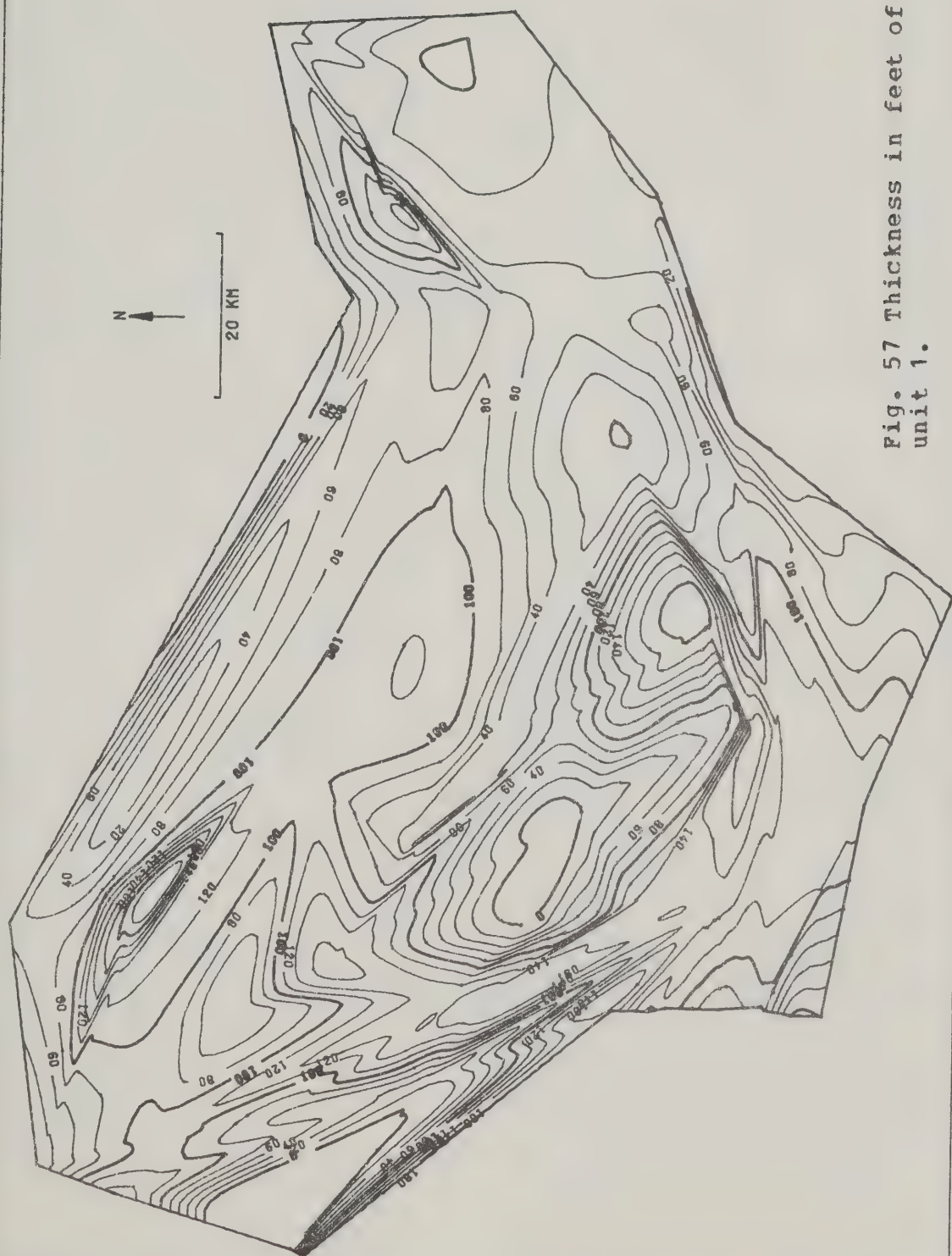


Fig. 57 Thickness in feet of till unit 1.



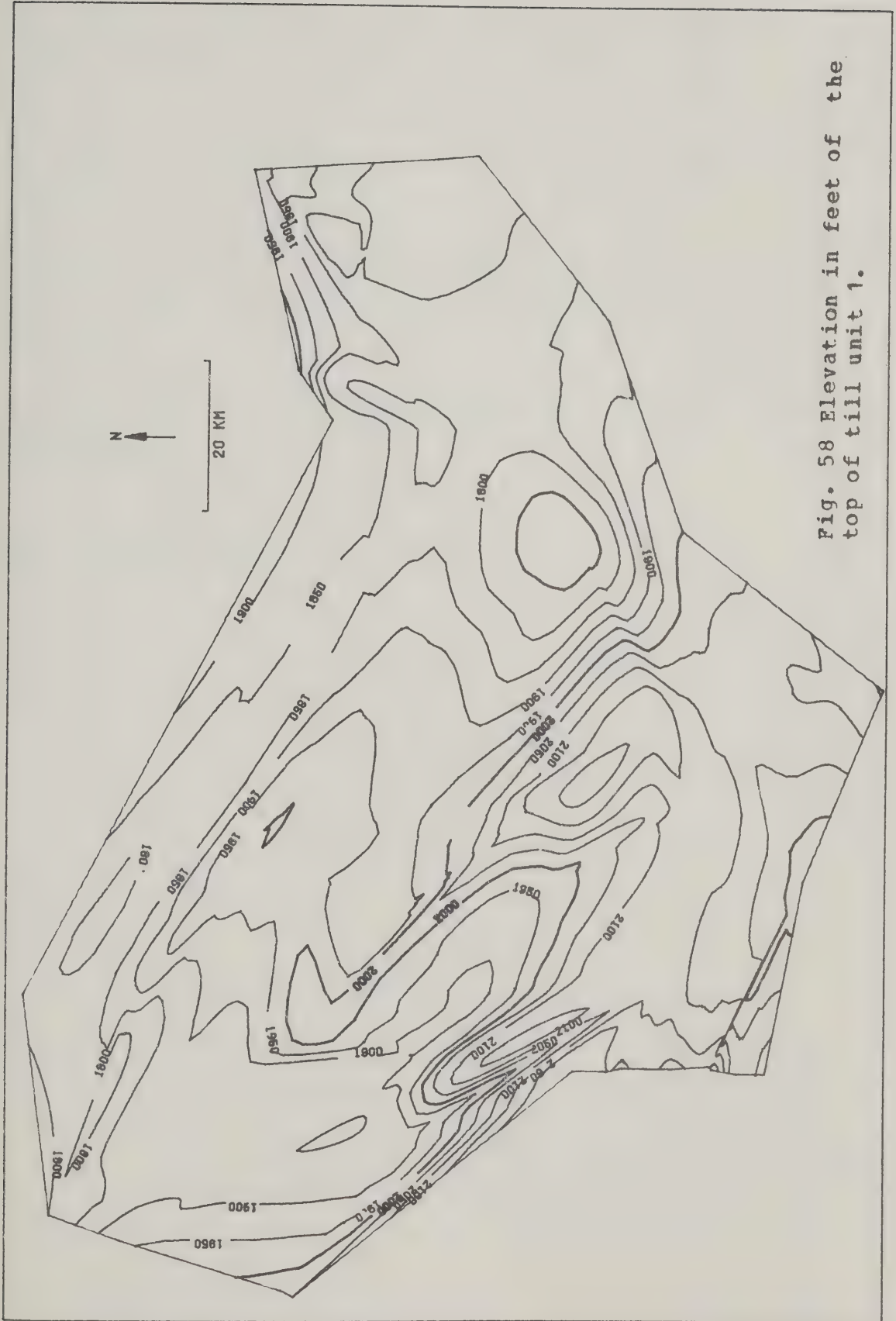


Fig. 58 Elevation in feet of the top of till unit 1.



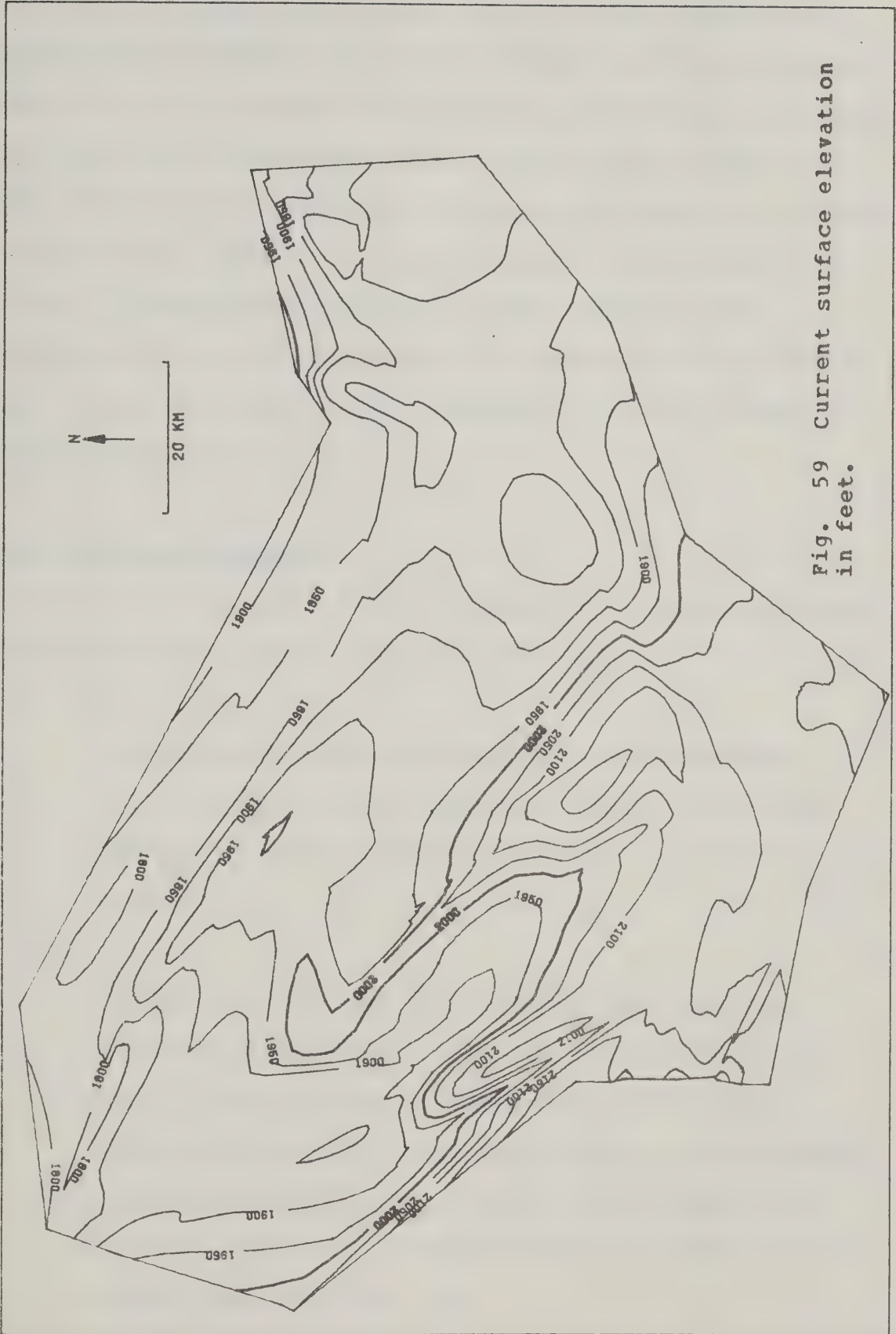


Fig. 59 Current surface elevation in feet.





of the most recent glaciation. Ellwood (1961) suggested a deglaciation sequence for the Vermillion area to the south that takes into account the meltwater channels in that area, as well as the late-stage remobilization of stagnant ice that produced the south-east trending flutings that are such a conspicuous feature on air-photographs. At an even later stage of deglaciation, Kocaoglu (1975) noted glacio-lacustrine deposits just below the junction of the Beverly and Kikino channels, and fluvio-glacial deposits slightly further upstream.

#### 9.7 Historical Summary

The glacial history of the area may be summarized with reference to the fence diagram of Fig. 49.

1/

The preglacial North Saskatchewan and Athabasca Rivers flowed east through the Beverly and Helina channels, joining in the eastern part of the map area.

2/

Glacial ice advanced into the map area from the north-east, depositing till unit 4 in the eastern part of the map area and disrupting the regional drainage. The North Saskatchewan River, being dammed, was diverted north, cutting the Kikino Channel. The Athabasca River was probably diverted north, out of the map area, at this time.



3/

The ice retreated to the east, and while the North Saskatchewan may have resumed its old course, abandoning the Kikino Channel, the Athabasca did not re-occupy the Helina Channel within the map area.

4/

After what may have been a relatively short time interval the ice readvanced into the area and deposited till unit 3. In the process of overriding the newly formed Kikino Channel bedrock thrusting occurred, completely blocking the Beverly Channel upstream from its junction with the Kikino Channel, forcing the North Saskatchewan to cut its present channel to the south. It is not known how far the ice extended to the west of the Kikino Channel.

5/

Following the retreat of the ice an indeterminate period of time elapsed before the arrival of the next ice sheet. No major drainage passed through the area.

6/

The new ice advance deposited the distinctive high-carbonate till unit 2 over the eastern half of the area, but apparently did not advance any further to the west.

7/

Following the retreat of this ice sheet an extensive period of time elapsed prior to the advance of the



final ice sheet, as indicated by the extensive weathered zone formed in many places on the land surface of that time (Section 7.1).

8/

The final ice advance from the east extended past the western boundary of the map area and deposited till unit 1. After the ice retreated this till formed much of the present topographic surface.

### 9.8 Correlation of Surficial Units

Christiansen (1968a,b, 1972a,b) and Whitaker and Christiansen (1972) classified the surficial deposits of central Saskatchewan on the basis of drill-hole information. The Empress, Sutherland and Saskatoon Groups established by them are described below.

The Empress Group is defined as those deposits between the bedrock surface and the lowest till, including both preglacial and glacial material. The Sutherland Group consists of the lowest major till and is distinguished from the overlying tills by possessing more clay and less sand, a generally lower electric log resistivity and a lower carbonate content, especially of dolomite. The Sutherland Group can sometimes be separated into an upper member with moderate and a lower member with very low carbonate content. Partial leaching of carbonates is sometimes found in the top 40 feet of the Sutherland Group.

The overlying Saskatoon Group is composed of the lower



Floral till and the upper Battleford till. These are not usually separable on the basis of carbonate content, but the Floral is frequently oxidized towards the top, indicating fairly extensive weathering, and is more compact and jointed (with oxide staining in the joints) than the overlying Battleford till which is more friable, soft and unjointed.

Table 11 shows the tentative correlation between Christiansen's work in Saskatchewan and the results of this study. On the basis of a lower carbonate content, lower electric-log resistivity, frequently lower sand values and the occasional separation into two units, the lower till of the Sand River area is correlated with Christiansen's Sutherland Group, although leaching of carbonates at the top is less obvious in the Sand River area. On the basis of the occurrence of an oxidized weathered zone directly below the upper till, the middle and upper tills of the Sand River area are correlated with the Floral and Battleford tills respectively. It should be noted that frequently in Christiansen's published cross-sections the Floral till does in fact have a rather higher carbonate content than the Battleford. In agreement also is the somewhat higher Mg content in the upper till of the Sand River area by comparison with the lower till.

With respect to carbonate values it should be noted that Christiansen's cross-sections tend to be parallel to ice-front positions and therefore approximately equidistant from the source of the erratic carbonate. Hence he does not





<u>Saskatchewan</u>		<u>Alberta</u>	
Christiansen (1968a,b, 1972a,b) Whitaker and Christiansen (1972)		This Study	
Saskatoon	Battleford Till	Upper Till	1
Group	— (Oxidized)	— (Oxidized)	2
	Floral Till	Middle Till	
Sutherland	— (Partially Leached)	Lower	3
	Upper member	Upper member	4
Group	Lower member	Till	
Empress		?	(Glacial)
Group		Sands and gravels	(Preglacial)
Bedrock		Bedrock	

Table 11. Tentative correlation of surficial units between Saskatchewan and Alberta.



specifically consider the decrease in carbonate content with distance from the source. In the Sand River area however, the main channels are roughly parallel to the ice flow direction and the effect of diluting the erratic carbonate with local shales is marked. This may well be due to relatively high erosion of local material due to the regional uphill slope facing the advancing ice.



## 10.0 CONCLUSIONS

### 10.1 Requirements for Stratigraphic Correlation

It has been shown in this work that it is possible to decipher the till lithostratigraphy and broad glacial history of an area from drill-hole information. The tools required are primarily the traditional ones of driller's and electric logs. The visual examination of cuttings or side-wall samples is not enough, and some compositional data are required. Different tills probably differ in composition in most areas because succeeding glaciations mask or expose different portions of bedrock, and the earlier tills themselves may be incorporated in the younger ones.

### 10.2 Choice of Compositional Measurements

Analysis of various compositional parameters indicates that many tills are remarkably homogeneous throughout their thickness, notwithstanding the occasional textural variation or sand lens that is apparent from the electric log. This is especially true of the silt and clay fraction in the present study. Lateral variation is less consistent, being affected by local sub-ice conditions and topography, although systematic dilution of erratic materials is evident away from their source. Compositional parameters should therefore be chosen from the fine rather than the coarse fraction of the till, since more individual grains will usually be analyzed in a procedure performed on a fine fraction. Total





sand content is frequently useful in separating till units within the drill-hole, as is the electric resistivity log. Compositional parameters in the fine material heavily reflect the ratio of erratic to local material, and in this study, as in others, calcium carbonate equivalent has proved to be the most concise summary of this, as well as being one of the easiest analyses to perform.

### 10.3 Sampling Methodology

On the basis of the plots in Appendix 4, a few empirical suggestions may be made on sample collection in the field. Several samples should be taken from each suspected till unit since a single sample may not be representative. When plotted in the fashion of Appendix 4, four samples will probably visually indicate the average composition of the till unit. The sampling decision would normally be made in the field on the basis of the previously-run electric log. Breaks in the general trend of the resistivity curve frequently denote a new lithologic unit and form a good basis for field sampling decisions. Samples every 5 feet are recommended for thin units, and every 20 or more feet for very thick tills. Deeper tills should be sampled at least as carefully as more shallow ones.



#### 10.4 Regional Stratigraphy

Four possible tills have been identified in the study area. Several of them may be regional in extent, as has been shown in Saskatchewan. It is suggested that Christiansen's (1968a,b, 1972a,b) nomenclature for surficial deposits (Empress, Sutherland and Saskatoon Groups; Floral and Battleford Formations) be used whenever the interpretation seems justified.

#### 10.5 Hydrology

Till stratigraphy has some benefit for estimating the hydrologic potential at various depths in an area. Sand or gravel layers are less likely to occur within a till than between tills, and if present are less likely to be laterally continuous. The zone between the lowest till and bedrock has a higher potential for water supply than the overlying tills, especially if associated with a bedrock valley. Inter-till sands are also fairly extensive, but usually tend to be less clean. Aquifers are most likely to be found in depressions in the surface on which they were formed - whether in a bedrock valley or a superimposed till surface. A basic understanding of the glacial stratigraphy and history should therefore facilitate the estimation of groundwater potential.



### 10.6 Analytic and Display Techniques

Techniques for bulk carbonate analysis, bulk chemistry of the fine fraction of till samples, and the computer display of contour maps have been developed for this study. In each case these developments appear to be tools the use of which extends beyond this work. The equipment developed is a step in the direction of automated laboratory carbonate analysis in both surficial geology and soil science. Bulk chemical analysis on fused till samples is currently dependent on the availability of an electron microprobe, but should soon be a viable technique on the more common, and less expensive, electron microscope when fitted with a suitable detector for energy-dispersive analysis. In terms of the rate of analysis of multiple elements and samples the method compares favourably with other available techniques.

Finally, the developments in automated contour mapping described here provide both facilities and economies not previously available. The approach outlined has several advantages in terms of flexibility, economy and the properties of the surface representation produced. Future enhancements could easily include slope calculations, automatic hill shading, storage of multiple surfaces (for example, geological contacts) and volumetric and areal calculations. In contrast with many methods the ideas used (i.e. breaking the region into triangles, obtaining the slopes at each data point and bending triangular plates to conform to these) are easy to comprehend, if not always easy



to implement, and hence the casual user should not receive any unpleasant surprises in the form of outrageous costs or surface shapes along with his desired map.





## REFERENCES

- Bannister, J.R., 1977. A magnetometer array study of polar magnetic substorms. Ph.D. thesis, Dept. of Geophysics, University of Alberta.
- Bengtsson, B.F. and Nordbeck, S., 1964. Construction of isarithms and isarithmic maps by computers. B.I.T. 4, 87-105.
- Borneuf, D., 1973. Hydrology of the Tawatinaw area, Alberta. Res. Council of Alberta Earth Science Rept. 72-11.
- Boulton, G.S. and Dent, D.L., 1974. The nature and rates of post-depositional changes in recently deposited till from south-east Iceland. Geografisker Annaler, v. 56, pp.121-134.
- Carlson, V.A., 1967. Bedrock Topography and Surficial Aquifers of the Edmonton district, Alberta. Res. Council of Alberta Earth Science Rept. 66-3.
- Carlson, V.A. and Currie, D.V., 1974. Bedrock Topography of the Vermillion map-area, Alberta, NTS 73E. Res. Council of Alberta Bedrock Topography Map. (See also Errata for this paper in Computer J., v.19, (Nov. 1976) p. 384.
- Carrigy, M.A., 1972. Lithostratigraphy of the Uppermost Cretaceous (Lance) and Paleocene strata of the Alberta plains. Res. Council of Alberta Bull. 27.
- Cattell, R.B., 1965a. Factor analysis - an introduction to essentials I - The purpose and underlying models. Biometrics, v. 21, pp. 190-215.
- Cattell, R.B., 1965b. Factor analysis - an introduction to essentials II - The role of factor analysis in research. Biometrics, v. 21, pp. 405-435.
- Cattell, R.B., 1966. The scree test for the number of factors. Multivariate Analysis in Behavioural Research 1, pp. 245-276.
- Christiansen, E.A., 1968a. A thin till in West-central Saskatchewan, Canada. Can. J. Earth Sci., v. 5, pp. 329-335.
- Christiansen, E.A., 1968b. Pleistocene stratigraphy of the Saskatoon area, Saskatchewan, Canada. Can. J. Earth Sci., v. 5, pp. 1167-1173.
- Christiansen, E.A., 1972a. Stratigraphy of the Fort Qu'Appelle vertebrate fossil locality, Saskatchewan.



Can. J. Earth Sci., v. 9, pp. 212-218.

- Christiansen, E.A., 1972b. Southern Saskatchewan, in Rutter N.W. and Christiansen, E.A. (eds.), Quaternary geology and geomorphology between Winnipeg and the Rocky Mountains. Field excursion C-22 guidebook, 24th International Geological Congress, Montreal, pp. 50-62.
- Christiansen, E.A. and Ross, W.C., 1971. Carbonate analysis of tills using a continuously-recording gasometer. Sask. Res. Council Rept. G71-5.
- Christiansen, E.A. and Whitaker, S.H., 1974. Geology and groundwater resources of the Waterhen River area (73-K), Saskatchewan. Sask. Res. Council Geology Division map no. 19.
- Cottafava, G. and LeMoli, G., 1969. Automatic Contour Map. Comm. Assoc. Comput. Mach., v. 12, pp. 386-389.
- Crain, I.K., 1970. Computer interpolation and contouring of two-dimensional data: A review. Geoexplor., v. 8, pp. 71-86.
- Day, P.R., 1965. Particle fractionation and particle size analysis, in Black, C.A. (ed.), Methods of soil analysis, Pt.1. Amer. Soc. Agron., Madison, Wisconsin.
- Dayhoff, M.O., 1963. A contour map program for X-ray crystallography. Comm. Assoc. Comput. Mach., v. 6, pp. 620-622.
- Douglas, R.J.W., 1968. Geological map of Canada. Geological Survey of Canada Map 1250A.
- Dreimanis, A., 1962. Quantitative gasometric determination of calcite and dolomite by using Chittick apparatus. J. Sed. Petrol., v. 5, pp. 520-529.
- Dreimanis, A., 1976. Tills: their origin and properties, in Leggett, R.F. (ed.), Glacial till. Roy. Soc. Can. Sp. Pub. 12.
- Dreimanis, A. and Vagners, U.J., 1969. Lithologic relation of till to bedrock, in Wright, H.E. Jr. (ed.), Quaternary geology and climate. National Academy of Science, Washington D.C., pp. 93-98.
- Dreimanis, A. and Vagners, U.J., 1971. Bimodal distribution of rock and mineral fragments in basal tills, in Goldthwait, R.P. (ed.), Till / A symposium. Ohio State University Press, pp. 237-250.



- Dyck, J.H. and Perkins, E.H., 1974. Heinze Gas-Volugraph: for recording gas evolved versus time. Can. J. Earth Sci., v. 11, pp. 565-569.
- Ellwood, R.B., 1961. Surficial geology of the Vermillion area, Alberta, Canada. Unpublished Ph.D. Thesis, University of Illinois, Urbana.
- Feuerbach, K.W., 1827. Grundriss zu analytischen Untersuchungen der dreieckigen Pyramide. Nurenburg.
- Frazer, J.Z., 1967. A computer fit to mass absorption coefficient data. Publication No. 67-29, Institute for the study of matter, University of California, La Jolla.
- Frenkel, Y. and Gill, D., 1975. An algorithm for contouring random data without gridding. Israeli J. Earth Sciences, v. 24, p. 56.
- Gillberg, G., 1965. Till distribution and ice movements on the northern slopes of the South Swedish Highlands. Geologiska Foreningens : Stockholm Forhandlingar, v. 86, pp. 433-484.
- Gillberg, G., 1967a. Further discussion of the lithological homogeneity of till. Geologiska Foreningens : Stockholm Forhandlingar, v. 89, pp. 22-49.
- Gillberg, G., 1967b. Distribution of different limestone material in till. Geologiska Foreningens : Stockholm Forhandlingar, v. 89, pp. 401-409.
- Gold, C.M., 1976a. Triangular element data structures. The University of Alberta Computing Services Users Applications Symposium Proceedings, Edmonton, pp. 43-54.
- Gold, C.M., 1976b. Surface estimation from irregularly spaced data points using an interpolant over irregular triangular domains. Geol. and Mineral. Assocs. Canada Program with Abstracts v. 1, p. 84.
- Gold, C.M., 1977. The practical generation and use of geographic triangular element data structures. Advanced Study Symposium on Topological Data Structures for Geographic Information Systems papers. Laboratory for Computer Graphics and Spatial Analysis, Harvard University, Oct. 1977.
- Gold, C.M., Charters, T.D. and Ramsden, J., 1977. Automated contour mapping using triangular element data structures and an interpolant over each triangular





- domain. Computer Graphics, v. 11, June 1977, pp. 170-175.
- Gold, C.M. and Smith, D.G.W., 1976. EDATA - a FORTRAN IV program for calculation of chemical compositions from spectra obtained by energy dispersive microbeam analysis. Internal Report, Department of Geology, University of Alberta, Edmonton.
- Goldthwait, R.P., 1971. Introduction to till today, in Goldthwaite, R.P. (ed.), Till / A symposium. Ohio State University Press.
- Green, R., 1972. Geological map of Alberta. Res. Council of Alberta, map 35.
- Harbaugh, J.W. and Merriam, D.F., 1968. Computer applications in stratigraphic analysis. John Wiley and Sons, New York.
- Harman, H., 1967. Modern factor analysis (2nd. ed.). University of Chicago Press.
- Hartigan, J.A., 1975. Clustering algorithms. John Wiley and Sons, New York.
- Heap, B.R., 1972. Algorithms for the production of contour maps over an irregular triangular mesh. N.P.L. report NAC 10.
- Heinrich, K.F.J., 1966. X-ray absorption uncertainty, in McKinley T.D., Heinrich K.F.J. and Wittry D.B. ( eds.), The electron microprobe, pp. 296-377. Wiley, New York.
- Irons, B.M., 1969. A conforming quartic triangular element for plate bending. Int. J. Num. Meth. Eng., v. 1, pp. 29-46.
- Kaiser, H.F., 1970. A second generation little jiffy. Psychometrika, v. 35, pp. 401-415.
- Kelly, T.K., 1966. Mass absorption coefficients and their relevance in electron-probe microanalysis. Trans. Inst. Min. Metall. v. 75, pp. B59-B73.
- Kocaoglu, S.S., 1975. Reconnaissance soil survey of the Sand River area (73L). Alberta Inst. Pedology Rept. no. S-74-34.
- Kramers, H.A., 1923. On the theory of X-ray absorption and the continuous X-ray spectrum. Phil. Mag., v. 46, pp. 836-871.



- Lawson, C.L., 1977. Software for C<sup>1</sup> surface interpolation. Jet Propulsion Laboratory Pub. 77-30, Aug. 1977.
- Le Breton, E.G., 1963. Groundwater Geology and Hydrology of east-central Alberta. Res. Council of Alberta Bull. 13.
- Lennox, D.H. and Carlson, V., 1967. Geophysical exploration for buried valleys on an area north of Two Hills, Alberta. Geophysics, v. 32, pp. 331-362.
- May, R.W. and Dreimanis, A., 1973. Differentiation of Glacial Till of southern Ontario based on their Cu, Zn, Cr and Ni geochemistry. Geol. Soc. Amer. Memoir 136, pp. 221-228.
- May, R.W. and Dreimanis, A., 1976. Compositional variability in tills, in Leggett, R.F. (ed.), Glacial till: an inter-disciplinary study. Roy. Soc. Can. Sp. Pub. 12.
- McConnell, R.G., 1886. Report on the Cypress Hills, Wood Mountain and adjacent country. Geol. Surv. Can. Ann. Rept., v. 1., pt. C., pp.1-78.
- McCrossan, R.G. and Glaister, R.P.(eds.), 1966. Geological history of western Canada. Alberta Soc. Petrol. Geol., Calgary.
- McLain, D.H., 1974. Drawing contours from arbitrary data points. Computer J., v. 17, pp. 318-324.
- McLain, D.H., 1976. Two dimensional interpolation from random data. Computer J., v. 19, pp. 178-181. See also Errata for this paper in Computer J., v. 19, p. 384.
- Nie, N.H., Hull, C.H., Jenkins, J.G., Steinbrenner, K. and Bent, D.H., 1975. Statistical Package for the Social Sciences, 2nd. ed., McGraw-Hill Inc., New York.
- Pawluk, S. and Bayrock, L.A., 1969. Some characteristics and physical properties of Alberta tills. Res. Council of Alberta Bull. 26.
- Perkins, E.H. and Dyck, J.H., 1973. Heinze Gas-Volugraph. Sask. Res. Council Rept.. P73-6.
- Peucker, T.K., 1972. Computer Cartography: Commission on College Geography resource paper no. 17, Assoc. Amer. Geographers, Washington D.C.
- Philibert, J., 1963. A method for calculating the absorption correction in electron-probe microanalysis, in Pattu H.H., Cosslett V.E. and Engstrom A. (eds.), X-ray optics and X-ray microanalysis, pp. 374-392. Academic



Press, New York.

- Pitteway, N.L.K., 1973. Computer graphics research in an academic environment. Datafair 73 Conference proceedings 2, pp. 471-478.
- Rao-Sahib, T.S. and Wittry, D.B., 1972. The X-ray continuum from thick targets, in Shinoda G., Kohra K. and Ichinokawa T. (eds.), Proc. 6th. Internat. Conf. X-ray Optics and Microanalysis, pp. 131-137. Univ. Tokyo Press, Japan.
- Reed, S.J.B. and Ware, N.G., 1972. Escape peaks and internal fluorescence in X-ray spectra recorded with lithium drifted silicon detectors. J. Phys. E.: Sci. Instrum. v. 5, pp. 582-584.
- Rhind, D., 1975. A skeletal overview of spatial interpolation techniques. Computer Applications, v. 2, pp. 293-309.
- Rucklidge, J.C., 1976. Instrumentation, in Smith, D.G.W. (ed.), Mineralogical Association of Canada Short Course Handbook, v. 1, pp. 1-44.
- Rutherford, R.L., 1937. Saskatchewan gravels and sands in central Alberta. Trans. Roy. Soc. Can., v. 31, pp. 81-95.
- Rutherford, R.L., 1941. Some aspects of glaciation in central and southwestern Alberta. Trans. Roy. Soc. Can., Section IV, pp. 115-124.
- St. Onge, D.A., 1972. Tawatinaw map-area geology, in Kjearsgaard, A.A., Reconnaissance soil survey of the Tawatinaw map sheet (83-I). Alberta Institute of Pedology Rept. S-72-29, pp. 93-95.
- Sampson, R.J., 1975. SURFACE II Graphics System. Series on Spatial Analysis no. 1, Kansas Geological Survey, Lawrence, Kansas.
- Sauer, E.K., 1974. Geotechnical implications of Pleistocene deposits in southern Saskatchewan, Canada. Can. Geotechnical J., v. 11, pp. 359-373.
- Savitzky, A. and Golay, M.J.E., 1964. Smoothing and differentiation of data by simplified least squares procedures. Anal. Chem., v. 36, pp. 1627-1638.
- SCA., 1970. MCS - Mapping Contouring System. Scientific Computer Applications Inc., Tulsa, Oklahoma.





- Schimann, K. and Smith, D.G.W., 1976. Energy dispersive whole rock analysis using an electron microprobe. Abstract, Canadian Probe Users' Group meeting, 21st. Mineral Ass. Can. Meeting, Edmonton.
- Shepard, D., 1968. A two-dimensional interpolation function for irregularly spaced data. Proc. 23rd. Natl. Conf., Assoc. Comput. Mach., pp. 517-524.
- Skinner, S.I.M. and Halstead, R.L., 1958. Note on rapid method for determination of carbonates in soils. Can. J. Soil Sci., v. 38, pp. 187-188.
- Skinner, S.I.M., Halstead, R.L. and Brydon, J.E., 1959. Quantitative manometric determination of calcite and dolomite in soils and limestones. Can. J. Soil Sci., v. 39, pp. 197-204.
- Smith, D.G.W., 1976. Quantitative energy dispersive microanalysis, in Smith, D.G.W. (ed.), Mineralogical Association of Canada Short Course Handbook, v. 1, pp. 63-106.
- Smith, D.G.W. and Gold, C.M., 1976. A scheme for fully quantitative energy dispersive microprobe analysis. Advances in X-ray Analysis, v. 19, pp. 191-201.
- Smith, D.G.W., Gold, C.M. and Tomlinson, D.A., 1975. The atomic number dependence of the X-ray continuum intensity and the practical calculation of background in energy dispersive electron microprobe analysis. X-ray Spectrom., v. 4, pp. 149-156.
- Sneath, P.H.A. and Sokal, R.R., 1973. Numerical taxonomy. Freeman and sons, New York.
- Springer, G., 1976. Correction procedures in electron-probe analysis, in Smith, D.G.W. (ed.), Mineralogical Association of Canada Short Course Handbook, v. 1, pp. 45-62.
- Stalker, A. MacS., 1968. Geology of the terraces at Cochrane, Alberta. Can. J. Earth Sci., v. 5, pp. 1455-1466.
- Steel, R.G.D. and Torrie, J.H., 1960. Principles and procedures in statistics. McGraw-Hill, New York.
- Turner, R.C., 1960. An investigation of the intercept method for determining the proportions of dolomite and calcite in mixtures of the two. I. Theoretical aspects of the rate of solution of dolomite when a number of crystals are present. Can. J. Soil Sci., v. 40, pp. 219-231.





- Turner, R.C. and Skinner, S.I.M., 1960. An investigation of the determining the proportions of dolomite and calcite in mixtures of the two. II. Experimental rates of solution of dolomite and calcite in samples of a number of crystals. *Can. J. Soil Sci.*, v. 40, pp. 232-241.
- Walters, R.F., 1969. Contouring by machine: A user's guide. *Amer. Assoc. Petrol. Geol. Bull.*, v. 53, pp. 2324-2340.
- Ware, N.G. and Reed, S.J.B., 1973. Background corrections for quantitative electron microprobe analysis using a lithium drifted silicon X-ray detector. *J. Phys. E.: Sci. Instrum.*, v. 6, pp. 286-288.
- Westgate, J.A., 1968. Surficial geology of the Foremost-Cypress Hills area, Alberta. *Res. Council Alberta Bull.* 22.
- Westgate, J.A., 1969. The Quaternary geology of the Edmonton area, Alberta, in Pawluk, S. (ed.), *Pedology and Quaternary Research*, University of Alberta Printing Department, Edmonton, pp. 129-151.
- Westgate, J.A., 1972. The Cypress Hills, in Rutter, N.W. and Christiansen, E.A. (eds.), *Quaternary geology and geomorphology between Winnipeg and the Rocky Mountains. Field excursion C-22 guidebook*, 24th International Geological Congress, Montreal, pp 50-62.
- Westgate, J.A., Kalas, L. and Evans, M.E., 1976. Geology of the Edmonton area. *Field trip C-8 guidebook*, Geological Association of Canada annual meeting, Edmonton.
- Whitaker, S.H. and Christiansen, E.A., 1972. The Empress Group in southern Saskatchewan. *Can. J. Earth Sci.*, v. 9, pp. 353-359.
- Willman, H.B., Glass, H.D. and Frye, J.C., 1963. Mineralogy of glacial tills and their weathering profiles in Illinois : Part I. Glacial tills. *Illinois Geological Survey Circular* 347.
- Winer, B.J., 1962. *Statistical principles in experimental design*. McGraw-Hill, New York.
- Wishart, D., 1975. *CLUSTAN 1C User Manual*. Computer Centre, University College, London.
- Yoon, T.N. and Vander Pluym, H., 1974. Buried channels in the Edmonton - Lac La Biche - Cold Lake area, Alberta. *Groundwater Development*
- Yoon, T.N., Gobeles, K. and Carlson, V., 1977. *Groundwater*



research in the Cold Lake oil sands area. Canada - Venezuela oil sands symposium 77 proceedings. (Awaiting publication). Branch Rept., Alberta Environment, Edmonton.

Zienkiewicz, O.C., 1971. The Finite Element Method in Engineering Science. McGraw Hill, London.



## APPENDIX 1

### Formal Laboratory Procedures

#### Initial Sample Preparation

Crush air-dried sample.

(Due to lengthy storage the sidewall samples are usually extremely hard. Consequently coarse crushing should be done with a jaw crusher followed by hand crushing with a rolling pin to avoid breaking rock fragments. Continue hand crushing until little new material passes through a #40 sieve.)

Sieve on #40 sieve (using "Sonic Sifter" for one minute at setting 6). Weigh and record the +40(A) and -40(B) fractions.

Wash +40 fraction on a #40 sieve and oven-dry.

Sieve on #10, #20 and #40 sieves using Sonic Sifter. Weigh and record the +10(C), +20(D) and +40(E) fractions.

Store -40 and total +40 fractions.

#### Grain Counts

Sieve +40 fraction on +10 and +20 sieves. Use -10, +20 fractions.

If necessary, partition sample by quartering until at least 200 grains are left in the pile. Categorize 200 grains as local shale or sandstone, acid igneous, basic igneous, carbonate or undifferentiated quartz. Record number of grains in each category, recombine and store +40 fraction.





### Grains Size Analysis

Sieve -40 fraction using #230 sieve until 5 to 10 gm of -230 fraction are obtained. Weigh and record. Place this sample in approximately 250 ml of Calgon solution (40 gm Calgon per litre distilled water), agitate and leave overnight to soak. 12 to 24 samples may be prepared at one time.

Distilled water should be kept at room temperature.

Wash sample into soil dispersion mixer with a small amount of distilled water. Mix at low speed for two minutes and high speed for one.

Wash sample back into jar with distilled water, make up to 400ml, agitate thoroughly and place on solid surface. Record starting time.

Calculate settling time for 2 micron fraction to fall 4 cm at room temperature using Stokes' Law. Assuming a particle density of 2.65,  $T$  in minutes =  $10^6 k^2$  where  $K=0.01365$  at 20°C.

At end of settling time insert the pipette to the 4 cm mark below the surface of the liquid. Siphons off 25 ml, place in a weighed bottle and oven-dry.

Record dry weight of the aliquot after subtracting the weight of the bottle and the weight of Calgon in a blank aliquot.

Wash remaining sample in a #230 sieve. Oven dry +230 fraction.

Sieve on #60, #140 and #230, sieves. Weigh and record



the -40 +60(H), -40 +140(I) and -40 +230(J) fractions.

Since it is necessary to use small samples and to extract some of the -230 fraction for further analysis, data reduction is as follows:

- a correction factor due to loss of -230 fraction,  
 $k = 1 + (A - E) / B$  is calculated.
- the fine fractions (H,I,J) are multiplied by k.
- the silt + clay weight =  $k(A - J)$ .
- the clay weight = volume of sample suspension x clay aliquot weight / aliquot volume.

### Carbonate Analysis

Power on the carbonate apparatus, and allow 10 minutes for the detector to stabilize. The recorder should be set at approximately 5 volts full scale and 8 minutes per inch.

Obtain 1 gm of sample by quartering the preserved -230 fraction and place it in the rinsed acrylic flask. Put 10 ml of 50% HCl in the reservoir, add a magnetic stirring bar, screw down the lid and allow the internal pressure to stabilize for a minute.

Invert flask to mix acid and sample. Place flask upright on magnetic stirrer set to about half speed.

Remove flask and unscrew lid when chart curve indicates no further pressure rise. Measure and record pressure rise due to the sample above the pressure in the uninverted sealed flask.

Run 5 calibrations of 200 mg oven-dried (105°C) calcium



carbonate reagent at least once a day, or if marked barometric changes occur.

Calculate % calcium carbonate equivalent for each sample assuming the average pressure change of a batch of calibrations to be due to 20% of calcium carbonate.

#### Bulk Chemistry - Sample Preparation

Obtain approximately 2 gm of preserved -230 fraction by quartering. Place in clean vial with 2 parts by volume of lithium tetraborate flux. Mix and pour into a clean crucible. Compact sample by tapping and invert onto graphite holder.

Place sample in image furnace, turn furnace up to 60 volts and allow sample to melt for 10 minutes. Remove sample, quench and break off tip of sample. Place in epoxy mount made by drilling 12 3/16" holes through a 1" disc of Araldite with a polyethylene ("Saran Wrap") film on one side. Record sample position. Fill mount with epoxy and leave 2 days to cure.

Grind on paper lap using #240 and #400 sandpaper.

Re-impregnate surface with 3M epoxy. Place in a vacuum for 10 minutes. Heat in oven at 110° C for 3 hours or overnight at 72°-75° C.

Grind briefly on #600 sandpaper, then wash it carefully with soap and water.

Impregnate mount with Geonite and wash with warm water.

Grind on paper using 6,3 and 1 micron diamond paste for



1, 1 and 3 hours respectively. Wash and re-impregnate with Geonite between paste changes.

Clean mount thoroughly, polish with .05 micron alumina for one minute, wash with warm water and soap.

Carbon coat mounts.

Bulk Chemistry - Analysis (University of Alberta  
Installation)

Place mounts in the ARL-AMX electron microprobe equipped with an Ortec lithium-drifted silica energy dispersive detector. Standards to be used are Kakanui kaersutite for Na, Mg, Al and Ca, quartz for Si, Hohenfels sanidine for K and ilmenite for Ti and Fe. These must be analyzed at the beginning of each run, and a calibration willemite standard should be processed at the beginning and end of the day's run. Instrument settings are 15 kV operating voltage, 300 nanoamps beam current and 8 microns scanning diameter for the beam. Beam current and probe current readings should be taken every hour with the Faraday cage.

Eight regions of each glass should be counted for 50 seconds each, for a total of 400 seconds per sample. Data is collected on the Texas Instruments 733 cassette tape terminal, and the saved data transmitted to the Amdahl 470/V6 computer for processing through the EDATA energy dispersive data-reduction program.





## APPENDIX 2

Calcium carbonate equivalence and grain size analyses for all analyzed samples.

The absolute values of the clay figures are considered to be unreliable due to disaggregation problems, although useful for comparison with neighbouring samples.

Samples are identified by drill hole number and depth in feet.



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand + Silt	%Sand, %Silt & Clay than 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve intervals)														
		10-20	10-40	10-60	10-140	10-230										
4342 5	2.5	4.47	10.83	21.69	38.60	46.71	92.18	100.00	1.15	46.71	45.47	7.82				
4342 10	6.9	4.12	10.31	19.14	34.05	40.32	95.66	100.00	2.17	40.32	55.34	4.34				
4342 15	7.8	3.46	10.48	24.42	50.05	61.03	84.95	100.00	0.62	61.03	23.92	15.05				
4342 20	6.4	4.96	12.82	16.63	21.53	23.20	90.02	100.00	7.50	23.20	66.83	9.99				
4342 25	6.6	4.17	11.23	21.39	39.01	46.47	89.52	100.00	1.90	46.47	43.05	10.48				
4342 30	7.2	4.50	12.26	22.67	41.64	45.76	91.16	100.00	2.20	45.76	45.40	8.84				
4342 35	8.6	3.46	9.65	20.17	38.53	45.71	90.01	100.00	2.35	45.71	44.29	9.99				
4342 45	7.5	3.50	10.38	21.54	37.95	45.00	86.92	100.00	2.79	45.00	41.92	13.08				
4342 50	8.3	3.65	9.97	20.99	38.46	46.03	92.13	100.00	4.24	46.03	46.10	7.87				
4342 55	5.5	3.39	10.62	21.96	41.40	49.18	90.19	100.00	2.66	49.18	41.01	9.81				
4342 60	3.8	3.27	10.74	10.85	34.05	38.99	92.25	100.00	1.02	38.99	53.25	7.75				
4342 135	7.2	1.81	5.26	11.88	26.46	33.52	94.21	100.00	0.50	33.52	60.69	5.79				
4342 140	6.3	1.97	6.48	12.47	23.87	30.19	91.95	100.00	0.62	30.19	61.76	8.05				
4342 145	6.7	1.33	4.47	9.53	19.79	25.74	88.98	100.00	1.32	25.74	63.25	11.02				
4342 150	7.6	0.82	2.88	6.36	13.50	19.48	84.73	100.00	0.25	19.48	65.25	15.27				
4342 155	6.5	1.37	4.20	9.34	18.59	25.73	91.26	100.00	0.50	25.73	65.53	8.74				
4352 10	4.9	3.52	10.42	20.53	38.76	46.72	91.39	100.00	1.50	46.72	44.68	8.61				
4352 20	6.1	3.55	9.80	18.58	34.59	41.01	89.85	100.00	3.23	41.01	48.84	10.15				
4352 30	5.1	3.03	9.09	17.92	34.89	42.35	88.78	100.00	1.30	42.35	46.43	11.22				
4352 40	7.5	4.10	11.02	21.11	39.96	45.69	93.23	100.00	1.43	45.69	47.55	6.77				
4352 50	7.2	3.42	10.56	21.43	39.23	46.15	94.18	100.00	2.48	46.15	48.03	5.82				
4352 55	6.6	3.07	9.18	18.34	37.44	43.64	93.18	100.00	2.18	43.64	49.54	6.82				
4352 60	8.4	3.00	8.83	17.41	40.78	48.28	93.62	100.00	1.79	48.28	45.34	6.38				



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -											%Sand + Silt	%Sand, %Coarser than 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve Intervals) - - - - -															
		10-20	10-40	10-60	10-140	10-230											
435E 70	6.3	4.04	11.49	22.58	41.00	50.30	93.85	100.00					50.30	43.55	6.15		
435E 80	4.8	3.57	10.46	20.52	37.43	44.10	88.49	100.00					44.10	44.39	11.51		
435E 90	2.8	2.95	8.04	17.34	31.49	38.32	89.44	100.00					38.32	51.12	10.56		
435E 100	3.3	1.91	7.11	15.88	29.66	36.24	93.98	100.00					36.24	57.74	6.02		
435E 110	4.4	2.38	7.82	20.36	34.67	40.00	90.18	100.00					40.00	50.19	9.82		
435E 120	3.4	2.49	8.17	17.23	31.46	38.35	92.93	100.00					38.35	54.58	7.07		
435E 130	2.1	2.16	7.17	16.08	32.60	39.44	93.33	100.00					39.44	53.89	6.67		
435E 140	2.3	3.64	10.47	18.83	33.64	40.22	94.97	100.00					40.22	54.75	5.03		
435E 160	2.9	2.32	7.74	17.30	34.53	41.79	95.76	100.00					41.79	53.97	4.24		
435E 170	2.5	2.19	7.25	16.63	35.59	42.31	92.78	100.00					42.31	50.47	7.22		
435E 172	2.7	3.35	10.54	23.15	40.71	46.26	92.37	100.00					46.26	46.11	7.63		
435E 180	3.0	1.19	5.14	14.56	31.47	37.78	91.77	100.00					37.78	53.99	8.23		
435E 190	2.8	2.26	7.99	16.19	34.32	41.85	93.49	100.00					41.85	51.64	6.51		
435E 200	6.0	3.59	10.20	17.37	30.13	35.89	88.44	100.00					35.89	52.55	11.56		
435E 210	4.7	4.27	11.66	19.88	35.51	42.14	96.08	100.00					42.14	53.94	3.92		
435E 220	3.4	2.30	7.22	15.59	31.05	37.62	87.36	100.00					37.62	49.74	12.64		
435E 250	2.7	1.13	4.42	10.77	23.78	29.53	82.59	100.00					29.53	53.06	17.41		
435E 252	2.6	1.44	5.16	11.15	25.14	34.30	83.22	100.00					34.30	48.91	16.78		
435E 260	3.1	1.61	5.22	12.02	24.40	30.80	83.48	100.00					30.80	52.69	16.52		
435E 270	2.6	0.27	1.16	5.24	25.60	35.07	70.33	100.00					35.07	35.26	29.67		
435E 290	6.1	2.99	8.29	15.71	29.47	36.39	88.91	100.00					36.39	52.52	11.09		
435E 310	5.2	2.58	7.82	14.01	27.96	34.65	88.44	100.00					34.65	53.80	11.56		
435E 320	5.4	2.56	6.09	12.14	25.00	32.42	83.86	100.00					32.42	51.44	16.14		
435E 330	2.6	7.51	14.08	20.01	30.17	35.29	87.70	100.00					35.29	52.42	12.30		
435E 340	4.5	2.02	5.61	12.32	25.51	32.78	87.09	100.00					32.78	54.31	12.91		





Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay	
		Sand (Sieve intervals) - - - - -															
		10-20	10-40	10-60	10-100	10-230											
4352 3#5	4.5	1.56	4.79	12.18	26.72	33.66						90.83	100.00	0.30	33.66	57.18	9.17
4362 10	4.2	3.49	9.05	16.64	32.47	40.12						90.29	100.00	6.00	40.12	50.17	9.71
4362 20	5.4	2.08	5.10	13.03	32.41	39.92						86.89	100.00	3.69	39.92	46.97	13.11
4362 30	5.7	3.34	9.43	18.85	35.64	42.91						87.61	100.00	4.20	42.91	44.70	12.39
4362 40	8.0	3.72	10.84	21.63	40.44	47.62						91.50	100.00	2.14	47.62	43.87	8.50
4362 50	6.5	5.43	12.88	23.18	41.47	48.80						89.29	100.00	10.67	48.80	40.49	10.71
4362 60	7.8	4.56	11.66	22.73	38.71	44.00						89.52	100.00	6.37	44.00	45.52	10.48
4362 70	6.8	4.09	11.72	22.26	40.62	47.50						89.26	100.00	3.19	47.50	41.77	10.74
4362 80	6.5	2.62	7.62	19.46	38.50	43.57						90.78	100.00	4.40	43.57	47.21	9.22
4362 85	6.1	3.54	9.99	21.32	40.05	46.16						87.09	100.00	2.50	46.16	40.93	12.91
4362 90	6.6	4.10	10.83	21.93	39.80	46.79						86.68	100.00	1.23	46.79	39.89	13.32
4362 95	6.8	3.89	10.73	20.92	37.77	45.29						88.26	100.00	1.78	45.29	42.97	11.74
4362 100	3.4	1.90	6.43	14.80	30.53	37.48						88.09	100.00	1.18	37.48	50.61	11.91
4362 110	3.4	2.15	6.96	14.63	30.65	36.29						88.66	100.00	0.60	36.29	52.37	11.34
4362 130	4.1	2.74	7.85	16.55	32.66	39.17						89.81	100.00	1.82	39.17	50.65	10.19
4362 150	3.6	2.23	6.99	14.59	31.72	37.49						88.89	100.00	1.00	37.49	51.40	11.11
4362 195	2.8	2.51	7.09	15.79	31.58	36.01						92.06	100.00	1.39	36.01	56.04	7.94
4372 5	7.0	2.16	7.18	15.90	34.32	40.87						88.41	100.00	0.90	40.87	47.55	11.59
4372 35	6.3	5.81	13.10	23.62	40.74	47.86						89.91	100.00	2.41	47.86	42.05	10.09
4372 45	8.1	4.16	8.00	12.44	32.58	39.07						88.83	100.00	10.49	39.07	49.77	11.17
4372 55	4.8	2.55	8.36	9.29	17.75	40.73						89.64	100.00	0.60	40.73	48.91	10.36
4372 65	4.4	4.83	13.38	21.95	37.57	43.72						92.07	100.00	6.83	43.72	48.36	7.93
4372 75	7.1	2.78	8.29	8.86	17.35	40.03						92.76	100.00	1.22	40.03	52.73	7.24



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															%Sand ↓ Silt & Clay	%Sand, %Coarser than 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-100	10-200	10-400	10-600	10-800	10-1000	10-1200	10-1400	10-1600	10-1800	10-2000						
437E 85	5.4	3.45	8.95	19.41	36.41	41.02	91.24	100.00	1.62	50.22	8.76	41.02	1.62	41.02	50.22	8.76					
437E 95	6.0	4.63	13.14	28.37	59.32	69.75	89.17	100.00	1.82	19.43	10.83	69.75	1.82	69.75	19.43	10.83					
437E 105	5.1	2.96	8.74	19.35	33.54	38.27	92.28	100.00	1.85	54.00	7.72	38.27	1.85	38.27	54.00	7.72					
437E 117	5.5	3.66	10.97	21.51	38.11	45.34	89.96	100.00	1.23	44.62	10.04	45.34	1.23	45.34	44.62	10.04					
437E 125	6.3	4.05	11.00	23.83	43.16	48.84	90.68	100.00	0.89	41.84	9.32	48.84	0.89	48.84	41.84	9.32					
437E 140	8.0	2.80	11.16	22.59	41.15	48.28	91.41	100.00	1.79	43.14	8.59	48.28	1.79	48.28	43.14	8.59					
437E 150	7.2	3.48	8.63	20.77	39.01	43.65	93.46	100.00	0.90	43.65	6.54	43.65	0.90	43.65	49.81	6.54					
437E 160	6.9	4.59	11.24	22.19	40.95	46.67	89.67	100.00	4.40	46.67	10.33	46.67	4.40	46.67	43.01	10.33					
437E 170	6.7	1.83	5.06	15.76	33.61	39.47	91.64	100.00	3.50	39.47	8.36	39.47	3.50	39.47	52.17	8.36					
437E 180	3.9	2.48	7.10	14.13	30.18	36.71	93.37	100.00	1.29	36.71	6.63	36.71	1.29	36.71	56.67	6.63					
437E 190	4.0	3.61	10.43	19.69	32.64	37.24	94.49	100.00	1.43	37.24	5.51	37.24	1.43	37.24	57.25	5.51					
437E 200	4.1	2.66	7.91	16.76	30.94	37.89	93.04	100.00	1.90	37.89	6.96	37.89	1.90	37.89	55.15	6.96					
437E 210	3.7	2.41	7.25	16.15	32.25	38.14	94.00	100.00	0.90	38.14	6.00	38.14	0.90	38.14	55.86	6.00					
437E 232	3.0	4.05	11.18	19.92	34.46	40.31	95.67	100.00	7.00	40.31	4.33	40.31	7.00	40.31	55.36	4.33					
437E 240	4.1	1.98	6.33	15.76	29.56	34.66	88.26	100.00	3.39	34.66	11.74	34.66	3.39	34.66	53.61	11.74					
437E 260	3.5	1.00	2.63	5.03	9.62	11.89	88.19	100.00	0.83	11.89	11.81	11.89	0.83	11.89	76.30	11.81					
439E 5	9.5	10.83	4.80	9.80	26.38	37.73	89.91	100.00	0.45	37.73	10.09	37.73	0.45	37.73	52.18	10.09					
439E 10	4.6	2.19	7.86	17.41	32.45	37.75	90.51	100.00	0.85	37.75	9.49	37.75	0.85	37.75	52.75	9.49					
439E 20	5.7	3.24	9.31	19.27	35.39	41.88	88.47	100.00	2.23	41.88	11.53	41.88	2.23	41.88	46.59	11.53					
439E 30	4.6	2.55	7.48	15.44	30.09	35.94	93.49	100.00	0.62	35.94	6.51	35.94	0.62	35.94	57.55	6.51					
439E 40	8.2	3.84	10.82	20.08	40.37	44.63	92.26	100.00	1.10	44.63	7.74	44.63	1.10	44.63	47.64	7.74					
439E 50	7.3	2.98	9.19	19.16	38.16	43.83	93.15	100.00	3.18	43.83	6.85	43.83	3.18	43.83	49.32	6.85					
439E 60	8.1	3.70	10.69	21.38	41.53	45.12	92.56	100.00	1.90	45.12	7.44	45.12	1.90	45.12	47.44	7.44					
439E 70	7.2	3.01	10.87	22.68	39.90	46.34	95.11	100.00	3.50	46.34	4.89	46.34	3.50	46.34	48.77	4.89					



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															
		Sand (Sieve intervals) - - - - -						%Sand + Silt	%Sand, %Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay				
10-20	10-40	10-60	10-140	10-230													
4392 80	8.7	4.22	11.64	21.99	40.53	46.52	93.17	100.00	2.20	46.52	46.66	6.83					
4392 90	7.2	3.83	10.55	23.28	43.17	47.91	91.96	100.00	1.95	47.91	44.05	8.04					
4392 100	7.8	3.92	11.01	24.52	43.44	47.65	93.76	100.00	3.80	47.65	46.11	6.24					
4392 108	12.6	4.40	11.34	21.14	39.23	44.36	93.05	100.00	7.20	44.36	48.69	6.95					
4392 118	2.6	2.91	8.38	16.83	37.62	45.98	90.34	100.00	0.92	45.98	44.36	9.66					
4402 10	4.3	3.88	9.87	16.63	28.27	32.16	84.69	100.00	1.92	32.16	52.53	15.31					
4402 20	7.9	4.58	12.15	20.85	49.03	56.22	90.45	100.00	2.69	56.22	34.23	9.55					
4402 30	4.7	3.64	9.82	17.15	38.56	43.27	87.72	100.00	1.30	43.27	44.45	12.28					
4402 40	6.2	2.66	8.22	15.27	33.17	38.94	94.00	100.00	1.00	38.94	55.07	6.00					
4402 50	3.8	3.36	9.66	20.11	37.40	43.22	90.52	100.00	1.50	43.22	47.29	9.48					
4402 60	8.4	3.49	9.78	18.61	37.85	42.35	93.83	100.00	2.03	42.35	51.48	6.17					
4402 70	2.7	2.34	6.49	11.92	25.95	31.09	88.72	100.00	0.29	31.09	57.63	11.28					
4402 80	4.5	1.51	4.58	8.99	34.51	41.22	91.82	100.00	0.43	41.22	50.60	8.18					
4402 90	3.4	2.49	7.72	15.55	32.71	37.71	93.18	100.00	1.30	37.71	55.47	6.82					
4402 100	4.7	2.49	7.14	14.36	29.07	33.75	91.73	100.00	0.50	33.75	57.98	8.27					
4402 110	4.2	2.35	6.89	12.52	27.29	31.37	96.29	100.00	0.62	31.37	64.92	3.71					
4402 120	4.3	2.97	7.68	15.01	28.25	33.79	91.94	100.00	0.62	33.79	58.15	8.06					
4402 150	4.5	2.65	8.04	16.10	32.40	36.96	94.55	100.00	2.10	36.96	57.59	5.45					
4402 170	2.0	2.42	7.37	14.57	31.19	35.02	97.54	100.00	1.78	35.02	62.52	2.46					
4402 180	1.1	2.23	7.01	14.11	30.50	36.73	90.49	100.00	1.00	36.73	53.77	9.51					
4402 190	2.3	2.44	6.68	12.72	27.23	30.77	98.32	100.00	0.30	30.77	67.55	1.68					
4402 210	2.4	1.97	6.57	14.07	32.01	36.04	97.91	100.00	0.48	36.04	61.88	2.09					
4402 219	2.7	2.32	7.16	15.30	30.90	37.43	92.53	100.00	0.32	37.43	55.10	7.47					
4402 230	4.3	2.00	5.68	11.10	26.10	29.90	95.91	100.00	0.79	29.90	66.02	4.09					



Hole/Depth	CaCO3	Cumulative - Percentages - - - - -										%Sand, %Silt & Clay than 10 mesh	%Sand	%Silt	%Clay
Equiv.	Sand (Sieve intervals) - - - - -														
		10-20	10-40	10-60	10-140	10-230	%Sand + Silt								
4402 240	5.3	4.01	8.88	15.90	33.13	39.14	87.03	100.00	6.23	39.14	47.88	12.97			
4402 250	4.5	2.31	5.73	11.41	29.51	33.25	94.51	100.00	1.22	33.25	61.26	5.49			
4402 256	2.5	1.85	5.39	11.78	29.04	32.42	94.12	100.00	0.90	32.42	61.71	5.88			
4552 5	4.8	1.82	5.97	13.30	30.90	33.64	91.61	100.00	0.48	33.64	57.98	8.39			
4552 10	3.7	2.90	3.93	8.67	19.03	20.54	89.98	100.00	1.50	20.54	69.44	10.02			
4552 15	5.1	1.32	4.30	10.02	22.90	26.34	88.61	100.00	0.39	26.34	62.27	11.39			
4552 20	5.0	1.49	4.57	9.80	24.36	27.70	87.60	100.00	0.50	27.70	59.89	12.40			
4552 35	13.4	3.46	8.17	14.27	29.95	34.37	90.53	100.00	4.98	34.37	56.16	9.47			
4552 50	8.5	4.06	11.70	21.05	36.90	40.76	92.62	100.00	3.75	40.76	51.86	7.38			
4552 65	13.8	3.01	6.93	12.06	26.80	31.82	92.40	100.00	3.76	31.82	60.57	7.60			
4552 80	15.0	3.38	7.86	13.32	26.09	29.73	90.97	100.00	2.64	29.73	61.24	9.03			
4552 95	10.9	3.17	8.34	15.56	31.59	35.56	88.60	100.00	2.62	35.56	53.05	11.40			
4552 110	11.1	3.81	9.15	16.76	32.02	36.05	89.23	100.00	1.57	36.05	53.18	10.77			
4562 5	3.9	1.38	4.85	10.85	25.78	31.35	87.10	100.00	1.78	31.35	55.75	12.90			
4562 15	4.8	0.20	1.12	2.26	4.78	5.46	88.33	100.00	0.31	5.46	82.87	11.67			
4562 25	3.6	2.26	7.24	16.55	37.61	42.84	91.48	100.00	0.78	42.84	48.65	8.52			
4562 35	5.1	1.66	5.40	12.33	27.20	31.48	86.68	100.00	0.26	31.48	55.21	13.32			
4562 40	6.3	2.11	5.86	12.94	27.53	32.42	86.48	100.00	0.50	32.42	54.06	13.52			
4562 45	5.8	1.97	5.71	13.04	29.26	35.05	91.89	100.00	0.90	35.05	56.85	8.11			
4562 55	6.5	1.12	3.74	8.65	18.29	20.14	91.37	100.00	0.50	20.14	71.23	8.63			
4562 65	5.1	1.52	3.99	8.14	16.70	18.62	89.59	100.00	1.37	18.62	70.97	10.41			
4562 75	6.5	1.24	3.53	8.09	16.61	18.50	90.87	100.00	0.30	18.50	72.37	9.13			
4562 85	6.7	1.86	4.75	10.53	22.43	25.01	90.79	100.00	7.62	25.01	65.79	9.21			





Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															%Sand + Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve intervals) - - - - -																			
		10-20	10-40	10-60	10-100	10-200															
4562 95	5.9	1.32	4.28	9.86	21.75	24.52	90.77	100.00	0.73	24.52	66.26	9.23									
4562 120	16.5	4.39	9.69	16.36	33.33	37.97	90.85	100.00	2.35	37.97	52.88	9.15									
4562 125	16.2	3.52	8.03	15.27	34.21	39.66	92.54	100.00	2.68	39.66	52.88	7.46									
4562 130	6.1	3.32	9.83	19.78	38.47	44.15	95.69	100.00	1.28	44.15	51.54	4.31									
4562 135	5.3	4.57	12.35	22.77	42.03	46.00	95.51	100.00	1.66	46.00	49.51	4.49									
4562 145	2.7	1.61	7.12	17.30	34.15	37.60	93.22	100.00	0.30	37.60	55.61	6.78									
4562 150	2.4	2.00	7.13	15.46	32.91	37.74	93.81	100.00	0.60	37.74	56.08	6.19									
4572 10	9.1	2.63	6.86	14.38	35.59	38.63	90.13	100.00	0.82	38.63	51.50	9.87									
4572 30	8.4	2.58	7.97	18.52	37.58	40.87	88.53	100.00	2.53	40.87	47.66	11.47									
4572 50	9.5	1.61	4.42	10.26	29.30	32.76	89.51	100.00	0.90	32.76	56.75	10.49									
4572 60	17.9	4.40	10.01	17.64	36.70	41.97	92.10	100.00	2.40	41.97	50.13	7.90									
4572 70	17.7	4.20	10.26	17.62	37.82	44.11	93.23	100.00	2.67	44.11	49.12	6.77									
4572 87	16.3	4.66	10.55	18.91	36.63	40.46	90.85	100.00	5.40	40.46	50.39	9.15									
4572 100	14.4	3.88	8.95	16.48	35.55	40.62	93.31	100.00	2.80	40.62	52.69	6.69									
4572 110	11.5	4.93	11.76	22.53	41.66	47.57	90.59	100.00	1.72	47.57	43.02	9.41									
4572 120	10.4	3.53	9.45	18.13	34.72	38.10	89.85	100.00	1.50	38.10	51.75	10.15									
4572 130	10.0	3.88	9.84	18.69	35.76	39.77	91.70	100.00	3.80	39.77	51.93	8.30									
4572 185	7.1	5.00	14.02	25.96	45.14	49.51	93.72	100.00	1.25	49.51	44.21	6.28									
4572 200	6.2	5.20	12.51	23.31	42.71	46.02	93.24	100.00	2.78	46.02	47.22	6.76									
4572 210	5.4	6.91	15.28	25.70	39.21	42.68	90.89	100.00	9.80	42.68	48.21	9.11									
4572 240	4.6	5.19	12.20	20.43	34.63	39.97	94.47	100.00	0.99	39.97	54.50	5.53									
4582 15	8.2	2.96	8.21	15.40	30.16	36.97	88.12	100.00	1.30	36.97	51.15	11.88									
4582 25	7.9	1.76	5.08	10.71	23.91	26.71	86.21	100.00	0.72	26.71	59.50	13.79									



Hole/Depth	CaCO3	Cumulative - Percentages - - - - -											%Sand, %Silt & Clay than 10 mesh	%Sand	%Silt	%Clay	
Equiv.	Sand (Sieve Intervals) - - - - -																
		10-20	10-40	10-60	10-140	10-230	%Sand ↑ Silt										
458E 80	6.8	4.90	12.62	21.75	38.05	41.96	90.09	100.00	4.40	41.96	48.13	9.91					
458E 185	7.2	4.05	11.53	22.02	42.27	46.93	93.56	100.00	0.41	46.93	46.63	6.44					
458E 225	6.9	5.08	13.01	22.69	42.68	46.57	93.81	100.00	10.53	46.57	47.24	6.19					
458E 235	6.4	1.00	4.22	12.53	35.79	41.63	92.67	100.00	0.20	41.63	51.04	7.33					
458E 255	9.1	4.51	11.03	19.62	34.81	37.86	87.91	100.00	2.20	37.86	50.05	12.09					
459E 10	7.3	2.18	6.69	13.34	29.79	33.90	87.34	100.00	1.38	33.90	53.43	12.66					
459E 40	8.8	1.61	4.99	11.25	28.56	31.95	88.68	100.00	0.82	31.95	56.73	11.32					
459E 60	9.1	1.12	3.32	7.27	22.10	26.41	92.02	100.00	0.72	26.41	65.61	7.98					
459E 70	17.7	4.30	9.81	17.72	35.99	40.18	90.77	100.00	6.26	40.18	50.58	9.23					
459E 100	9.8	2.24	5.53	12.26	22.72	25.52	85.35	100.00	3.48	25.52	59.83	14.65					
461E 5	5.4	1.79	5.76	12.31	28.62	34.20	87.30	100.00	1.19	34.20	53.10	12.70					
461E 10	4.4	1.81	6.05	12.71	29.39	35.91	84.76	100.00	0.80	35.91	48.85	15.24					
461E 15	3.0	2.14	6.39	13.72	29.08	34.94	84.19	100.00	0.60	34.94	49.25	15.81					
461E 20	3.8	2.14	6.69	13.89	31.15	35.07	90.36	100.00	0.50	35.07	55.29	9.64					
461E 25	5.9	3.82	7.82	15.74	28.61	34.14	88.26	100.00	5.43	34.14	54.13	11.74					
461E 30	4.9	1.58	22.20	28.89	40.78	45.52	86.71	100.00	1.15	45.52	41.19	13.29					
461E 35	6.0	1.61	4.78	10.45	25.59	29.84	88.70	100.00	0.58	29.84	58.86	11.30					
461E 40	5.4	1.94	5.28	12.76	24.69	29.31	82.02	100.00	0.63	29.31	52.71	17.98					
461E 45	5.7	1.70	4.97	12.03	28.24	32.68	87.20	100.00	1.20	32.68	54.52	12.80					
461E 50	3.8	1.67	5.75	12.68	29.98	35.88	90.20	100.00	0.10	35.88	54.32	9.80					
461E 65	4.4	1.78	4.86	12.28	22.38	25.68	87.17	100.00	0.53	25.68	61.49	12.83					
461E 69	5.9	4.12	11.00	24.16	42.23	46.36	92.11	100.00	3.31	46.36	45.75	7.89					
461E 75	6.5	2.15	8.14	16.81	31.38	36.01	91.54	100.00	2.23	36.01	55.54	8.46					



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand + Silt	%Sand finer than No. 10 mesh	%Silt	%Clay
		Sand (Sieve Intervals) - - - - -													
		10-20	20-40	40-60	60-100	100-200	200-400	400-800	800-1600	1600-3200	3200-6400				
462E 5	4.3	1.39	10.30	33.46	65.19	68.45	91.06	100.00	100.00	100.00	0.03	62.45	22.61	8.94	
462E 20	5.1	1.54	5.24	11.87	27.75	31.98	91.51	100.00	100.00	100.00	0.29	31.98	59.53	8.49	
462E 35	5.0	1.93	6.05	11.76	29.77	35.13	91.40	100.00	100.00	100.00	3.50	35.13	56.27	8.60	
462E 50	16.0	3.71	8.87	15.31	30.04	35.34	91.20	100.00	100.00	100.00	4.78	35.34	55.86	8.80	
462E 60	10.1	2.90	7.49	14.06	27.59	31.70	86.81	100.00	100.00	100.00	4.10	31.70	55.11	13.19	
462E 75	13.0	5.07	11.38	19.33	36.19	40.84	91.37	100.00	100.00	100.00	4.99	40.84	50.52	8.63	
462E 90	9.1	3.28	8.12	15.24	27.22	31.26	90.47	100.00	100.00	100.00	2.49	31.26	59.21	9.53	
463E 10	5.1	2.75	7.70	15.07	32.19	35.76	85.26	100.00	100.00	100.00	1.10	35.76	49.50	14.74	
463E 20	4.8	1.59	5.52	12.91	29.09	32.40	88.71	100.00	100.00	100.00	0.98	32.40	56.31	11.29	
463E 25	5.0	1.69	5.93	13.41	27.88	34.17	83.13	100.00	100.00	100.00	0.82	34.17	48.96	16.87	
463E 30	5.0	2.03	6.02	13.97	27.72	33.52	85.95	100.00	100.00	100.00	1.03	33.52	52.43	14.05	
463E 40	6.2	2.20	5.65	11.75	24.03	27.48	87.41	100.00	100.00	100.00	0.48	27.48	59.94	12.59	
463E 50	5.4	2.11	6.00	13.50	28.11	31.49	87.94	100.00	100.00	100.00	1.08	31.49	56.45	12.06	
463E 60	5.0	1.44	3.00	6.25	22.88	27.61	91.55	100.00	100.00	100.00	1.03	27.61	63.94	8.45	
463E 70	6.1	1.87	5.51	14.16	27.29	32.26	89.65	100.00	100.00	100.00	1.20	32.26	57.40	10.35	
463E 75	6.2	1.65	4.66	10.26	17.41	20.48	83.58	100.00	100.00	100.00	0.69	20.48	63.10	16.42	
463E 80	5.4	4.63	11.29	20.96	37.05	42.00	92.28	100.00	100.00	100.00	5.78	42.00	50.29	7.72	
463E 85	5.0	3.44	9.61	18.67	34.06	38.80	92.16	100.00	100.00	100.00	2.29	38.80	53.36	7.84	
463E 90	5.6	2.79	7.76	16.10	30.03	34.21	90.84	100.00	100.00	100.00	0.71	34.21	56.63	9.16	
463E 95	5.9	2.95	7.55	15.44	27.67	31.92	91.73	100.00	100.00	100.00	0.47	31.92	59.81	8.27	
463E 100	6.1	3.01	7.52	13.57	28.83	30.77	88.31	100.00	100.00	100.00	1.45	30.77	57.54	11.69	





Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand, %Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230	%Sand + Silt									
496E 210	5.7	2.49	6.69	12.17	30.29	36.57	93.41	100.00	0.49	36.57	56.84	6.59				
496E 250	1.9	2.38	6.97	12.61	23.53	27.56	93.72	100.00	1.23	27.56	66.16	6.28				
496E 290	1.8	2.94	9.36	19.46	40.65	45.26	92.19	100.00	0.31	45.26	46.93	7.81				
496E 310	1.6	2.47	8.83	20.39	38.85	43.62	91.54	100.00	0.85	43.62	47.91	8.46				
496E 330	2.3	1.52	4.85	10.44	24.51	28.07	94.10	100.00	0.25	28.07	66.03	5.90				
496E 345	2.4	1.89	5.64	11.79	25.74	30.34	90.04	100.00	1.79	30.34	59.70	9.96				
496E 365	3.7	2.42	6.76	14.87	28.93	34.76	91.51	100.00	0.50	34.76	56.76	8.49				
497E 30	11.3	4.07	9.22	16.83	30.78	34.50	88.91	100.00	2.98	34.50	54.41	11.09				
497E 40	10.1	3.69	9.05	17.08	30.51	35.98	86.90	100.00	1.01	35.98	50.92	13.10				
497E 50	7.9	2.10	6.07	11.91	30.21	33.79	89.80	100.00	0.79	33.79	56.01	10.20				
497E 80	5.5	1.91	5.84	12.85	29.88	32.58	87.98	100.00	0.20	32.58	55.40	12.02				
497E 90	3.7	1.31	4.16	9.23	27.22	31.03	90.06	100.00	0.83	31.03	59.02	9.94				
497E 100	5.7	2.35	6.94	15.37	32.24	37.77	88.49	100.00	2.00	37.77	50.73	11.51				
497E 110	5.4	1.80	6.41	14.92	34.63	37.64	89.88	100.00	1.40	37.64	52.24	10.12				
497E 120	6.4	3.37	6.63	13.28	25.69	30.96	89.36	100.00	0.40	30.96	58.41	10.64				
497E 130	12.7	3.69	9.00	17.53	32.06	35.64	88.80	100.00	2.00	35.64	53.17	11.20				
497E 140	10.8	3.08	9.11	16.69	29.84	34.90	86.99	100.00	3.20	34.90	52.10	13.01				
497E 150	12.0	5.59	12.02	18.12	36.60	42.59	93.37	100.00	9.11	42.59	50.78	6.63				
497E 160	12.2	4.76	11.04	18.00	33.51	38.66	94.12	100.00	3.40	38.66	55.46	5.88				
497E 170	12.4	3.92	9.24	16.63	34.79	38.31	92.45	100.00	2.23	38.31	54.14	7.55				
497E 220	5.3	2.41	6.63	13.64	29.42	32.60	86.56	100.00	1.40	32.60	53.96	13.44				
497E 230	5.7	5.20	9.72	15.67	23.53	26.68	86.04	100.00	3.40	26.68	59.36	13.96				
497E 240	5.0	0.95	2.75	5.84	13.91	15.64	85.95	100.00	0.35	15.64	70.31	14.05				
497E 250	6.8	3.26	9.63	19.21	41.87	48.79	93.21	100.00	0.85	48.79	44.42	6.79				



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															%Sand + Silt & Clay 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve intervals) - - - - -																		
		10-20	10-40	10-60	10-140	10-230														
497E 260	5.9	4.00	10.44	19.62	39.25	44.41						92.24	100.00	1.17	22.41	47.83	7.76			
497E 270	6.4	2.23	6.70	14.11	29.97	36.45						90.78	100.00	0.45	36.45	54.33	9.22			
497E 280	5.2	1.13	2.61	6.18	14.68	16.67						84.14	100.00	0.82	16.67	67.47	15.86			
497E 290	6.1	1.42	4.15	9.17	17.71	20.76						86.74	100.00	0.49	20.76	65.98	13.26			
497E 295	5.5	1.44	4.81	10.38	22.61	25.46						84.77	100.00	0.29	25.46	64.31	10.23			
498E 15	4.2	1.63	5.61	12.98	34.37	37.62						85.48	100.00	1.13	37.62	47.86	14.52			
498E 25	4.9	1.81	6.26	19.15	34.25	35.89						93.31	100.00	1.45	35.89	57.42	6.69			
498E 40	4.1	1.61	5.87	13.32	30.84	34.18						91.92	100.00	1.10	34.18	57.74	8.08			
498E 55	4.2	1.55	5.47	12.37	27.95	30.66						89.81	100.00	0.80	30.66	59.16	10.19			
498E 65	4.3	2.61	6.88	13.87	28.96	35.42						87.07	100.00	4.53	35.42	51.65	12.93			
498E 70	5.5	1.73	5.60	12.55	28.32	32.27						90.86	100.00	1.01	32.27	58.59	9.14			
498E 100	11.2	3.11	7.80	14.69	28.29	31.61						89.33	100.00	2.07	31.61	57.72	10.67			
498E 110	9.6	2.41	6.56	12.83	26.02	28.88						96.15	100.00	2.13	28.88	67.27	3.85			
498E 120	13.0	3.08	8.03	14.71	28.59	32.97						86.00	100.00	3.79	32.97	53.04	14.00			
498E 130	9.8	2.16	6.35	12.78	23.68	27.30						90.14	100.00	0.80	27.30	62.84	9.86			
498E 140	12.8	3.03	7.61	13.21	26.34	29.19						89.65	100.00	1.50	29.19	60.46	10.35			
498E 150	9.3	4.87	10.04	16.91	31.45	35.99						91.41	100.00	2.80	35.99	55.41	8.59			
498E 160	5.0	2.08	6.62	15.71	31.64	37.71						86.41	100.00	1.61	37.71	48.70	13.59			
498E 170	8.0	3.47	9.07	15.92	27.70	32.55						89.16	100.00	1.39	32.55	56.61	10.84			
498E 180	8.1	2.54	6.56	12.30	25.29	29.36						83.91	100.00	1.39	29.36	54.55	16.09			
498E 190	13.4	18.92	35.25	49.35	60.55	63.65						93.84	100.00	6.99	63.65	30.19	6.16			
498E 200	11.8	7.74	20.43	42.42	61.03	63.99						93.14	100.00	1.15	63.99	29.15	6.86			
498E 250	10.8	3.84	9.38	17.07	31.74	35.01						82.54	100.00	3.13	35.01	47.53	17.46			
498E 265	6.3	3.73	8.75	17.81	32.81	35.60						89.07	100.00	2.40	35.60	53.47	10.93			



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															%Sand + Silt & Clay	%Sand, %Silt & Clay 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve intervals) - - - - -																			
		10-20	20-40	40-60	60-100	100-140	140-230														
498E 270	7.2	2.94	6.28	12.60	26.01	31.31	89.41	100.00	1.20	31.31	58.10	10.59									
498E 275	6.8	2.62	7.63	17.03	37.21	41.20	88.33	100.00	0.79	41.20	47.14	11.67									
498E 280	7.1	1.42	4.52	9.26	18.79	22.70	88.98	100.00	0.79	22.70	66.28	11.02									
498E 290	6.6	2.08	5.74	12.10	24.23	27.09	86.00	100.00	0.82	27.09	58.92	14.00									
706E 50	5.6	3.92	10.80	20.97	37.83	44.86	90.61	100.00	3.42	44.86	45.75	9.39									
706E 60	3.7	3.51	10.87	12.82	15.06	17.01	82.22	100.00	1.20	17.01	65.21	17.78									
706E 70	6.1	3.98	11.66	23.53	41.01	47.29	96.24	100.00	1.59	47.29	48.95	3.76									
706E 80	6.2	1.98	6.81	14.15	34.29	45.87	91.71	100.00	1.50	45.87	45.84	8.29									
706E 100	7.3	3.36	10.89	20.96	37.78	45.11	90.07	100.00	1.10	45.11	44.96	9.93									
706E 110	7.2	4.54	12.46	23.33	37.80	42.41	88.17	100.00	10.08	42.41	45.76	11.83									
706E 120	5.9	3.04	9.43	19.42	42.19	49.81	89.74	100.00	3.52	49.81	39.93	10.26									
706E 125	7.0	4.19	11.82	22.70	37.99	43.44	87.80	100.00	7.70	43.44	44.35	12.20									
706E 130	7.5	3.76	11.08	22.05	39.52	45.35	89.69	100.00	2.90	45.35	44.34	10.31									
706E 200	7.9	3.53	10.59	20.61	35.38	41.91	91.43	100.00	0.80	41.91	49.53	8.57									
706E 210	4.3	2.23	6.82	13.77	25.30	32.08	88.06	100.00	0.80	32.08	55.98	11.94									
706E 220	5.1	2.90	7.86	14.83	26.88	34.34	87.79	100.00	0.71	34.34	53.45	12.21									
706E 230	4.3	2.79	8.10	15.19	29.10	35.46	86.00	100.00	1.40	35.46	50.53	14.00									
706E 240	3.5	2.05	7.31	15.07	29.51	35.91	88.81	100.00	0.92	35.91	52.90	11.19									
706E 250	3.7	1.10	3.71	7.53	26.83	41.38	86.68	100.00	0.30	41.38	45.30	13.32									
706E 260	5.6	3.53	9.23	16.22	30.87	37.52	87.20	100.00	0.30	37.52	49.68	12.80									
706E 270	3.3	2.09	6.67	14.34	29.63	37.55	87.26	100.00	1.40	37.55	49.71	12.74									
706E 280	2.9	2.21	6.97	15.40	29.50	36.71	87.33	100.00	1.49	36.71	50.62	12.67									
706E 290	3.6	1.53	5.68	12.85	28.10	36.48	88.40	100.00	0.42	36.48	51.92	11.60									



Hole/Depth	CaCO <sub>3</sub>	Cumulative - Percentages										%Sand	%Silt	%Clay
		Equiv. - - - Sand (Sieve Intervals)										%Sand, %Coarser Silt & Clay than 10 mesh	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230								
706E 325	3.4	2.41	8.34	15.97	29.08	37.04	89.69	100.00	6.11	37.04	52.65	10.31		
706E 345	2.8	2.20	7.50	15.30	29.90	39.41	90.35	100.00	0.89	39.41	50.94	9.65		
706E 348	3.4	1.76	6.61	15.37	28.57	35.44	84.56	100.00	3.30	35.44	49.12	15.44		
706E 360	2.9	2.40	7.79	17.98	31.70	38.30	83.76	100.00	2.88	38.30	45.46	16.24		
709E 10	6.9	1.19	3.39	15.90	33.90	41.24	85.58	100.00	0.49	41.24	44.35	14.42		
709E 20	5.2	5.56	17.20	23.50	32.62	37.50	78.40	100.00	1.91	37.50	40.90	21.60		
709E 30	6.1	3.55	10.00	20.70	36.91	43.76	87.75	100.00	2.80	43.76	43.99	12.25		
709E 40	7.2	3.23	9.55	20.82	37.19	43.99	90.44	100.00	1.28	43.99	46.44	9.56		
709E 50	7.4	4.05	10.65	21.87	38.58	45.33	89.23	100.00	4.00	45.33	43.91	10.77		
709E 60	7.0	3.33	9.50	20.09	35.27	41.85	90.06	100.00	1.71	41.85	48.21	9.94		
709E 70	7.5	3.91	10.22	22.11	38.79	45.46	89.18	100.00	6.70	45.46	43.72	10.82		
709E 80	9.1	3.30	9.05	19.64	34.15	40.28	85.86	100.00	4.09	40.28	45.57	14.14		
709E 90	9.6	2.89	8.01	17.63	30.91	36.32	89.58	100.00	0.60	36.32	53.26	10.42		
709E 100	2.9	3.57	10.66	22.20	37.04	43.59	87.27	100.00	2.30	43.59	43.68	12.73		
709E 110	3.0	3.50	10.62	22.83	38.22	45.16	88.13	100.00	2.10	45.16	42.97	11.87		
709E 120	2.8	4.11	12.03	24.13	41.04	48.16	88.32	100.00	2.39	48.16	40.15	11.68		
709E 130	3.7	2.70	10.13	22.68	39.67	47.62	92.68	100.00	2.90	47.62	45.05	7.32		
709E 140	3.6	4.05	11.63	25.37	42.24	49.72	94.97	100.00	5.18	49.72	45.25	5.03		
709E 150	3.6	2.65	8.76	19.64	33.65	40.02	92.96	100.00	1.20	40.02	52.94	7.04		
709E 160	3.8	2.06	5.59	12.68	22.42	36.17	86.94	100.00	2.79	36.17	50.77	13.06		
709E 170	3.5	3.02	8.74	20.08	35.73	42.01	89.65	100.00	2.30	42.01	47.65	10.35		
710E 5	6.6	2.19	7.85	17.95	39.21	46.45	86.82	100.00	0.70	46.45	40.36	13.18		





Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand + Silt	%Sand, Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve intervals) - - - - -															
		10-20	10-40	10-60	10-140	10-230											
710E 90	7.0	3.39	11.01	21.05	35.65	42.04						85.20	100.00	0.30	42.04	43.16	14.80
710E 95	6.9	6.83	15.53	28.51	43.95	50.08						94.95	100.00	1.50	50.08	44.87	5.05
710E 110	6.9	3.92	11.10	22.26	39.01	45.58						91.55	100.00	1.31	45.58	45.97	8.45
710E 115	6.8	4.28	11.71	23.10	39.38	45.97						90.23	100.00	4.32	45.97	44.26	9.77
710E 120	5.8	3.31	9.95	21.94	38.03	44.54						91.91	100.00	0.39	44.54	47.37	8.09
710E 125	7.0	4.63	12.88	25.40	40.49	46.98						92.82	100.00	2.53	46.98	45.84	7.18
710E 130	6.6	3.97	11.19	22.43	37.93	44.20						93.14	100.00	3.70	44.20	48.94	6.86
710E 135	6.9	4.83	12.51	23.92	39.58	46.12						93.53	100.00	1.05	46.12	47.42	6.47
710E 140	7.0	3.91	11.31	22.86	38.01	44.33						92.77	100.00	5.18	44.33	48.44	7.23
710E 145	6.9	5.29	11.99	23.41	38.98	45.95						94.34	100.00	9.20	45.95	48.39	5.66
710E 150	6.6	4.60	12.92	25.18	40.51	46.99						91.58	100.00	11.50	46.99	44.59	8.42
710E 155	6.5	2.80	9.24	19.87	38.41	45.73						92.59	100.00	2.13	45.73	46.86	7.41
710E 160	7.0	4.39	11.37	21.99	37.99	44.91						89.82	100.00	4.23	44.91	44.91	10.18
710E 165	6.8	3.75	10.99	22.60	38.18	45.06						91.36	100.00	0.90	45.06	46.30	8.64
710E 244	7.4	9.30	16.83	28.81	45.20	51.47						92.05	100.00	1.88	51.47	40.58	7.95
710E 273	5.5	5.45	10.44	20.16	44.46	58.41						86.04	100.00	12.91	58.41	27.63	13.96
713E 60	2.8	1.01	2.95	6.56	13.77	20.37						82.18	100.00	0.79	20.37	61.81	17.82
713E 70	6.2	3.45	9.23	21.68	36.87	42.76						88.08	100.00	3.09	42.76	45.31	11.92
713E 80	7.9	2.08	6.76	17.40	31.03	38.22						90.16	100.00	0.75	38.22	51.94	9.84
713E 90	5.0	1.19	3.46	10.91	19.49	23.23						84.83	100.00	1.91	23.23	61.59	15.17
713E 100	7.0	3.05	9.29	23.15	38.15	44.59						92.40	100.00	3.06	44.59	47.82	7.60
713E 110	5.9	2.42	6.80	16.37	29.77	36.33						88.96	100.00	1.81	36.33	52.63	11.04
713E 120	8.1	1.55	4.70	12.25	23.39	27.30						90.03	100.00	1.19	27.30	62.73	9.97



Bore/Depth	Caco3 - - - - - Cumulative - Percentages - - - - -												
	Equiv. - - - - - Sand (Sieve intervals) - - - - -												
	10-20	10-40	10-60	10-140	10-230	%Sand ↑ Silt & Clay	%Sand, %Silt & Clay 10 mesh	%Coarser than 10 mesh	%Sand	%Silt	%Clay		
7142 5	12.6	3.30	8.25	15.45	27.52	32.58	88.47	100.00	1.20	32.58	55.89	11.53	
7142 10	11.7	3.14	7.44	14.13	26.74	31.81	86.92	100.00	1.85	31.81	55.11	13.08	
7142 15	13.2	3.34	9.68	17.42	29.95	35.87	87.07	100.00	2.14	35.87	51.20	12.93	
7142 20	11.6	3.66	8.49	15.45	26.58	31.65	87.07	100.00	2.39	31.65	55.42	12.93	
7142 25	13.8	3.03	6.51	14.57	26.01	32.21	87.99	100.00	2.10	32.21	55.78	12.01	
7142 30	10.7	3.88	7.42	11.96	19.97	23.76	87.00	100.00	1.31	23.76	63.24	13.00	
7142 35	13.1	3.20	7.67	15.21	26.41	31.87	90.00	100.00	4.35	31.87	58.13	10.00	
7142 40	12.2	3.36	7.94	14.67	27.08	32.80	90.37	100.00	1.20	32.80	57.58	9.63	
7142 45	11.7	3.34	8.64	15.74	27.91	33.29	91.03	100.00	2.91	33.29	57.73	8.97	
7142 50	12.9	3.96	8.07	15.99	28.40	33.81	90.79	100.00	2.51	33.81	56.98	9.21	
7142 55	11.3	1.73	4.87	13.82	26.20	29.37	89.78	100.00	0.21	29.37	60.41	10.22	
7142 60	16.6	4.62	11.30	21.74	43.97	50.17	89.75	100.00	2.69	50.17	39.57	10.25	
7142 65	11.2	2.83	7.04	15.04	24.95	29.73	86.95	100.00	0.20	29.73	57.22	13.05	
7142 70	12.3	3.86	8.79	15.53	28.08	34.05	88.73	100.00	2.01	34.05	54.68	11.27	
7142 75	12.6	4.02	9.18	16.85	28.55	34.15	87.85	100.00	2.30	34.15	53.70	12.15	
7142 80	11.9	3.07	7.32	13.31	23.61	28.27	86.11	100.00	2.50	28.27	57.84	13.89	
7142 85	16.0	2.18	5.42	10.90	19.36	24.43	89.58	100.00	1.90	24.43	65.15	10.42	
7142 90	12.7	2.99	7.28	14.89	24.74	28.54	85.05	100.00	1.17	28.54	56.50	14.95	
7142 95	12.3	4.61	9.66	16.62	28.14	33.34	86.08	100.00	12.32	33.34	52.74	13.92	
7142 100	6.5	4.18	11.27	23.05	39.44	46.79	91.52	100.00	2.59	46.79	44.73	8.48	
7142 105	6.6	3.90	10.29	21.37	36.16	42.77	89.13	100.00	0.40	42.77	46.36	10.87	
7142 110	7.0	3.84	10.22	21.17	36.48	43.00	92.99	100.00	2.25	43.00	49.99	7.01	
7142 115	5.5	1.96	5.66	11.48	19.83	24.42	91.23	100.00	0.57	24.42	66.81	8.77	
7142 120	5.5	2.66	7.00	12.89	23.30	28.18	89.22	100.00	0.81	28.18	61.04	10.78	
7142 125	5.1	2.20	5.84	13.05	21.99	26.38	86.89	100.00	0.61	26.38	60.50	13.11	



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -														%Sand finer than 10 mesh	%Silt	%Clay
		Sand (Sieve intervals) - - - - -																
		10-20	20-40	40-60	60-100	100-200	200-400	400-800	800-1600	1600-3200	3200-6400	6400-12800	12800-25600	25600-51200	51200-102400			
714E 130	6.2	2.37	6.34	12.54	22.84	28.11	86.18	100.00	1.18	28.11	58.06	13.82						
714E 135	5.1	2.18	7.54	13.61	23.81	28.81	86.82	100.00	0.91	28.81	58.01	13.18						
714E 140	4.9	2.30	5.84	12.03	21.85	26.20	89.15	100.00	1.40	26.20	62.95	10.85						
714E 145	6.2	1.76	6.41	12.35	22.00	26.84	87.07	100.00	1.09	26.84	60.23	12.93						
714E 150	5.7	1.93	5.37	10.71	20.72	26.51	80.69	100.00	1.21	26.51	54.18	19.31						
714E 155	4.9	2.52	6.63	13.32	24.16	29.23	84.47	100.00	0.37	29.23	55.25	15.53						
714E 160	6.0	2.22	6.34	12.82	24.73	29.71	83.14	100.00	0.53	29.71	53.43	16.86						
714E 165	5.1	1.70	5.92	15.43	27.26	32.36	81.87	100.00	0.01	32.36	49.51	18.13						
714E 170	2.8	2.03	5.37	13.56	24.93	29.84	81.53	100.00	0.21	29.84	51.68	18.47						
714E 175	3.9	1.81	5.65	15.33	27.77	33.00	83.92	100.00	0.65	33.00	50.92	16.08						
714E 180	4.3	3.88	8.44	17.44	29.44	33.69	85.40	100.00	9.00	33.69	51.71	14.60						
714E 185	3.9	4.35	6.05	13.09	23.19	28.44	88.94	100.00	0.81	28.44	60.50	11.06						
714E 190	3.7	2.40	6.59	13.70	23.72	29.01	90.42	100.00	1.00	29.01	61.41	9.58						
714E 195	3.9	2.73	7.25	15.32	25.90	31.05	88.82	100.00	2.89	31.05	57.77	11.18						
714E 200	2.8	1.21	5.49	13.11	23.56	29.35	89.37	100.00	0.75	29.35	60.02	10.63						
715E 10	6.1	1.79	5.81	13.41	26.81	32.63	83.64	100.00	0.10	32.63	51.01	16.36						
715E 20	7.0	3.68	10.30	20.69	37.39	44.32	88.92	100.00	2.53	44.32	44.60	11.08						
715E 30	5.9	3.57	10.13	18.67	33.98	40.34	87.99	100.00	1.62	40.34	47.64	12.01						
715E 40	7.2	3.93	10.65	20.69	37.15	43.41	90.35	100.00	2.73	43.41	46.94	9.65						
716E 10	7.9	2.14	6.97	16.30	31.65	38.11	83.43	100.00	0.68	38.11	45.32	16.57						
716E 20	5.2	2.34	6.12	13.80	25.04	31.11	87.24	100.00	4.80	31.11	56.13	12.76						
716E 30	6.5	3.64	6.52	12.63	24.77	31.04	85.76	100.00	0.39	31.04	54.72	14.24						
716E 40	7.5	3.37	10.03	21.92	36.13	42.24	90.23	100.00	0.60	42.24	47.99	9.77						





Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															
		Sand (Sieve intervals) - - - - -															
		10-20	10-40	10-60	10-140	10-230	%Sand + Silt	%Sand, %Silt & Clay 10 mesh	%Sand	%Silt	%Clay						
7162 50	7.4	3.33	9.89	22.43	36.30	42.34	91.56	100.00	1.49	42.34	49.22	8.44					
7162 60	6.6	3.63	9.89	19.46	34.46	40.52	90.50	100.00	2.12	40.52	49.99	9.50					
7162 70	7.4	3.36	11.58	23.06	36.86	42.81	92.49	100.00	1.59	42.81	49.68	7.51					
7162 80	6.5	3.67	9.80	19.36	34.28	40.00	85.25	100.00	1.84	40.00	45.26	14.75					
7162 90	6.6	3.73	10.05	19.95	33.74	39.79	88.09	100.00	1.10	39.79	48.29	11.91					
7162 100	7.7	4.11	11.56	22.06	38.73	45.59	89.17	100.00	2.58	45.59	43.58	10.83					
7162 110	7.7	1.78	5.84	18.33	41.53	48.08	88.41	100.00	0.92	48.08	40.32	11.59					
7162 120	5.2	2.97	8.15	14.96	26.19	30.86	83.08	100.00	2.34	30.86	52.22	16.92					
7162 130	2.6	0.79	2.55	4.61	7.58	9.62	69.28	100.00	0.40	9.62	59.66	30.72					
7162 140	4.8	3.03	8.42	15.99	29.11	35.30	90.67	100.00	0.30	35.30	55.38	9.33					
7162 150	5.5	3.76	10.39	19.49	32.26	38.13	89.11	100.00	5.54	38.13	50.97	10.89					
7162 160	7.3	3.69	10.26	21.03	35.36	41.61	87.52	100.00	3.09	41.61	45.91	12.48					
7162 170	5.8	5.24	13.51	22.35	38.67	44.89	87.97	100.00	2.52	44.89	43.09	12.03					
7162 180	6.6	4.27	11.69	21.57	38.77	45.73	89.58	100.00	3.80	45.73	43.84	10.42					
7162 190	5.8	3.36	9.38	17.94	32.24	36.95	85.59	100.00	0.69	36.95	48.64	14.41					
7162 200	6.2	3.27	9.22	17.31	32.90	38.55	82.32	100.00	1.62	38.55	43.77	17.68					
7162 210	10.6	3.81	9.86	18.17	32.46	38.40	89.71	100.00	4.21	38.40	51.31	10.29					
7162 220	9.4	2.90	8.01	15.40	28.82	34.32	85.51	100.00	3.05	34.32	51.18	14.49					
7162 230	8.1	3.21	9.73	22.18	37.82	43.30	85.10	100.00	0.68	43.30	41.80	14.90					
7162 240	2.7	2.96	9.26	19.39	31.25	35.61	83.91	100.00	0.88	35.61	48.30	16.09					
7162 250	3.7	3.59	9.78	22.02	33.00	38.50	88.89	100.00	0.50	38.50	50.39	11.11					
7162 260	4.4	4.65	12.11	21.32	35.16	38.82	90.19	100.00	0.90	38.82	51.37	9.81					
7162 270	3.9	4.30	10.71	21.17	34.03	39.65	91.36	100.00	0.85	39.65	51.71	8.64					
7162 280	4.6	4.29	11.96	22.02	36.61	43.05	89.81	100.00	2.11	43.05	46.75	10.19					
7162 290	4.4	4.63	11.51	20.94	34.42	40.08	88.79	100.00	3.25	40.08	48.71	11.21					



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand + Silt	%Sand, Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230											
716E 310	7.4	0.59	3.62	8.46	13.38	16.18	83.91	100.00	0.80	16.18	67.73	16.09					
716E 349	4.0	3.12	9.53	21.94	35.71	42.22	86.04	100.00	0.72	42.22	43.82	13.96					
716E 360	4.3	2.73	8.09	18.74	33.34	40.41	91.48	100.00	1.00	40.41	51.07	8.52					
717E 15	4.8	5.86	14.00	22.85	38.80	46.27	87.42	100.00	4.01	46.27	41.15	12.58					
717E 20	7.4	8.47	19.55	31.82	47.66	52.99	90.26	100.00	6.80	52.99	37.27	9.74					
717E 25	8.2	5.61	13.00	23.36	39.27	46.28	90.33	100.00	10.00	46.28	44.04	9.67					
717E 30	7.3	4.45	11.30	21.75	35.75	41.91	92.85	100.00	3.63	41.91	50.94	7.15					
717E 35	9.3	3.93	10.87	21.01	37.59	44.35	95.09	100.00	3.54	44.35	50.74	4.91					
717E 40	7.0	4.73	12.45	23.01	38.44	45.27	91.15	100.00	3.90	45.27	45.88	8.85					
717E 60	7.9	4.44	13.04	24.85	43.34	50.99	95.54	100.00	2.95	50.99	44.55	4.46					
717E 120	4.7	3.01	7.98	15.43	27.60	34.19	89.81	100.00	0.77	34.19	55.62	10.19					
717E 130	4.2	4.12	9.50	18.16	28.88	34.46	88.18	100.00	10.10	34.46	53.73	11.82					
717E 140	5.7	1.82	5.81	12.40	23.74	29.94	90.02	100.00	0.61	29.94	60.07	9.98					
717E 150	5.4	1.25	10.65	18.58	30.41	36.96	90.56	100.00	1.89	36.96	53.60	9.44					
717E 160	7.7	0.50	1.74	3.75	7.34	9.36	79.64	100.00	0.0	9.36	70.28	20.36					
717E 170	4.6	1.81	6.52	14.75	29.47	38.10	88.52	100.00	1.00	38.10	50.42	11.48					
717E 210	4.5	2.33	5.72	12.01	21.49	27.64	86.88	100.00	2.69	27.64	59.24	13.12					
718E 10	2.7	0.99	3.61	9.14	19.57	25.95	82.57	100.00	0.22	25.95	56.62	17.43					
718E 20	4.0	3.09	9.52	20.50	34.57	41.39	92.61	100.00	0.94	41.39	51.22	7.39					
718E 30	1.1	0.0	0.67	2.31	7.46	23.88	87.05	100.00	0.0	23.88	63.17	12.95					
718E 40	7.9	1.85	5.63	11.04	21.64	28.68	87.96	100.00	1.03	28.68	59.28	12.04					
718E 50	4.1	1.98	5.23	14.15	27.14	34.40	88.43	100.00	2.00	34.40	54.03	11.57					
718E 60	4.3	2.55	7.81	16.53	29.72	36.41	90.35	100.00	1.34	36.41	53.94	9.65					



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -												%Sand + Silt	%Sand, Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230	10-425	10-600	10-840	10-1060	10-1490	10-2000	10-2970						
718E 70	3.8	2.69	7.89	16.61	30.11	36.76	40.08	44.77	48.86	52.44	55.51	58.08	60.16	91.13	100.00	0.85	36.76	54.38	8.87
718E 80	4.3	4.27	10.79	19.27	32.79	40.08	44.77	48.86	52.44	55.51	58.08	60.16	62.24	84.77	100.00	0.65	40.08	44.69	15.23
718E 90	4.0	2.92	8.43	18.44	31.88	37.93	41.49	44.77	47.93	50.00	52.07	54.14	56.21	91.49	100.00	1.12	37.93	53.56	8.51
718E 100	3.8	1.85	6.26	13.43	26.24	31.42	36.59	41.76	46.93	52.10	57.27	62.44	67.61	91.58	100.00	0.57	31.42	60.16	8.42
718E 110	3.2	2.62	8.71	20.56	35.24	42.48	49.72	56.96	64.20	71.44	78.68	85.92	93.16	91.72	100.00	0.77	42.48	49.24	8.28
718E 120	3.6	2.76	8.46	17.78	32.89	39.79	46.69	53.59	60.49	67.39	74.29	81.19	88.09	92.23	100.00	1.71	39.79	52.45	7.77
718E 130	2.9	2.89	9.39	19.91	35.55	42.28	48.91	55.54	62.17	68.80	75.43	82.06	88.69	91.63	100.00	1.42	42.28	49.35	8.37
718E 140	2.5	2.61	8.43	19.32	35.23	42.63	49.03	55.43	61.83	68.23	74.63	81.03	87.43	88.43	100.00	0.78	42.63	45.80	11.57
718E 150	3.1	1.30	4.26	10.07	19.22	24.24	29.26	34.28	39.29	44.31	49.33	54.35	59.37	84.28	100.00	0.13	24.24	60.04	15.72
718E 160	4.4	2.08	6.05	12.68	25.08	32.40	39.72	47.04	54.36	61.68	69.00	76.32	83.64	87.19	100.00	0.81	32.40	54.78	12.81
718E 170	3.9	3.87	6.16	12.26	25.47	32.04	38.61	45.18	51.75	58.32	64.89	71.46	78.03	83.94	100.00	1.70	32.04	51.90	16.06
718E 180	4.7	2.00	6.28	14.52	29.23	37.97	46.71	55.45	64.19	72.93	81.67	90.41	99.15	84.34	100.00	1.99	37.97	46.37	15.66
718E 190	4.4	2.72	8.88	20.39	34.76	40.39	46.02	51.65	57.28	62.91	68.54	74.17	79.80	87.09	100.00	3.20	40.39	46.71	12.91
718E 200	4.7	2.86	8.97	17.84	31.85	37.13	42.41	47.69	52.97	58.25	63.53	68.81	74.09	90.55	100.00	0.60	37.13	53.43	9.45
718E 210	4.8	3.20	8.45	18.64	33.35	38.44	43.53	48.62	53.71	58.80	63.89	68.98	74.07	89.13	100.00	7.21	38.44	50.69	10.87
719E 10	6.3	2.15	7.00	14.83	28.88	35.46	42.04	48.62	55.20	61.78	68.36	74.94	81.52	81.36	100.00	0.88	35.46	45.89	18.64
719E 20	5.8	3.62	9.60	18.60	33.44	39.45	45.46	51.47	57.48	63.49	69.50	75.51	81.52	82.86	100.00	4.25	39.45	43.41	17.14
719E 30	7.2	3.54	9.82	19.79	34.58	40.64	46.69	52.74	58.79	64.84	70.89	76.94	82.99	90.34	100.00	1.49	40.64	49.71	9.66
719E 40	8.6	4.82	12.14	22.74	38.58	45.36	52.14	58.92	65.70	72.48	79.26	86.04	92.82	89.01	100.00	8.75	45.36	43.65	10.99
719E 50	8.8	3.89	11.17	22.36	39.13	46.59	54.05	61.51	68.97	76.43	83.89	91.35	98.81	88.32	100.00	2.02	46.59	41.73	11.68
719E 60	3.2	3.94	11.40	22.44	38.16	44.84	51.52	58.20	64.88	71.56	78.24	84.92	91.60	86.39	100.00	2.01	44.84	41.55	13.61
719E 70	3.3	2.48	7.41	15.95	27.67	34.55	41.43	48.31	55.19	62.07	68.95	75.83	82.71	82.73	100.00	2.42	34.55	48.18	17.27
719E 80	3.6	2.91	8.68	17.78	30.54	36.67	42.79	48.91	55.03	61.15	67.27	73.39	79.51	91.17	100.00	1.40	36.67	54.49	8.83
719E 90	4.6	5.35	14.29	25.53	41.98	48.95	55.92	62.89	69.86	76.83	83.80	90.77	97.74	93.89	100.00	2.63	48.95	44.95	6.11



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages																%Sand + Silt	%Sand, %Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																						
		10-20	10-40	10-60	10-140	10-230	10-425	10-600	10-840	10-1060	10-1490	10-2000	10-2970	10-4750	10-7500	10-11900	10-20000																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																												
719E 100	4.3	3.75	11.38	28.53	38.53	44.62	92.06	100.00	0.98																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																																				</

48.44

28.27

48.88

50.36

50.19

50.29

37.91

48.44

23.30





Hole/Depth	Caco3      - - - - - Cumulative - Percentages - - - - -												
	Equiv.      - - - - - Sand (Sieve intervals) - - - - -												
	10-20	10-40	10-60	10-140	10-230	%Sand + Silt	%Sand, %Coarser Silt & Clay 10 mesh	%Sand	%Silt	%Clay			
725E 10	5.4	2.10	6.79	13.82	25.63	31.22	88.58	100.00	1.09	51.22	57.36	11.42	
725E 15	7.4	1.92	5.45	14.29	29.82	37.53	91.94	100.00	0.75	37.53	54.40	8.06	
725E 20	7.0	3.16	7.86	13.43	23.73	28.56	84.21	100.00	6.51	28.56	55.64	15.79	
725E 25	7.0	2.04	6.18	12.80	24.62	30.43	84.24	100.00	1.81	30.43	53.82	15.76	
725E 30	6.6	4.47	9.49	16.53	28.34	33.92	86.10	100.00	2.09	33.92	52.19	13.90	
725E 35	7.4	1.90	5.59	10.86	23.05	30.53	81.70	100.00	0.50	30.53	51.16	18.30	
725E 40	7.5	2.77	7.72	14.25	27.44	34.18	84.33	100.00	2.00	34.18	50.16	15.67	
725E 45	9.1	2.23	5.14	9.47	18.80	23.09	85.82	100.00	1.31	23.09	62.73	14.18	
725E 60	7.3	1.70	6.48	15.92	35.84	44.19	85.86	100.00	0.61	44.19	41.67	14.14	
725E 65	6.7	2.79	6.87	11.88	24.13	29.71	88.29	100.00	1.51	29.71	58.58	11.71	
725E 70	5.1	1.76	5.31	11.05	20.30	24.05	88.47	100.00	0.91	24.05	64.42	11.53	
725E 75	8.1	2.55	7.07	13.54	26.94	32.67	83.17	100.00	1.00	32.67	50.50	16.83	
725E 80	6.3	1.43	4.13	8.08	18.31	23.41	82.81	100.00	0.31	23.41	59.40	17.19	
725E 85	6.1	1.52	4.73	9.09	18.23	23.05	76.03	100.00	0.45	23.05	52.98	23.97	
725E 90	4.8	1.37	4.38	8.52	15.70	19.71	86.31	100.00	1.32	19.71	66.61	13.69	
725E 95	7.3	12.92	19.81	29.41	46.58	51.96	84.14	100.00	0.75	51.96	32.18	15.86	
725E 100	7.5	2.24	8.61	18.28	33.42	38.91	91.84	100.00	1.05	38.91	52.93	8.16	
725E 105	6.8	3.72	9.30	19.05	32.89	37.81	89.04	100.00	16.18	37.81	51.23	10.96	
725E 108	6.4	9.38	17.26	25.20	49.76	55.33	84.13	100.00	15.62	55.33	28.80	15.87	
725E 115	6.8	2.08	5.86	11.90	21.91	25.64	83.56	100.00	0.65	25.64	57.92	16.44	
725E 120	5.5	2.24	5.77	11.97	23.09	27.91	83.69	100.00	2.14	27.91	55.78	16.31	
725E 125	7.8	2.23	5.55	11.24	21.50	25.58	85.44	100.00	0.36	25.58	59.85	14.56	
725E 130	6.3	2.37	6.24	12.14	22.82	28.04	86.28	100.00	2.35	28.04	58.24	13.72	
725E 136	7.0	4.87	11.60	21.62	34.79	40.39	78.18	100.00	1.48	40.39	37.79	21.82	
725E 140	8.3	5.67	12.95	22.14	37.40	43.27	86.66	100.00	1.00	43.27	43.39	13.34	



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand + Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230	10-230	10-230	10-230	10-230	10-230					
725E 145	4.7	1.29	3.96	9.17	17.95	21.58					80.58	100.00	0.95	21.58	59.00	19.42
725E 150	7.7	2.85	8.03	15.98	29.03	34.38					84.61	100.00	2.80	34.38	50.23	15.39
725E 155	7.0	4.14	8.96	16.55	29.38	35.17					83.79	100.00	4.31	35.17	48.62	16.21
725E 160	7.8	1.76	5.32	11.79	23.00	28.79					85.51	100.00	0.59	28.79	56.72	14.49
725E 165	6.5	4.21	9.84	16.38	30.85	39.06					85.41	100.00	3.95	39.06	46.35	14.59
725E 170	8.5	2.09	6.30	12.63	27.01	34.66					85.73	100.00	0.42	34.66	51.08	14.27
725E 175	6.6	2.12	5.32	10.54	20.31	25.44					87.13	100.00	0.69	25.44	61.68	12.87
725E 180	5.7	0.74	1.81	3.75	7.29	10.05					81.92	100.00	0.13	10.05	71.86	18.08
725E 190	6.3	5.20	10.49	17.91	29.69	35.65					85.34	100.00	14.30	35.65	49.69	14.66
725E 195	8.4	2.67	7.45	13.76	26.39	32.19					88.64	100.00	0.63	32.19	56.45	11.36
725E 200	6.5	2.28	6.54	12.47	24.68	30.59					88.60	100.00	5.59	30.59	58.01	11.40
725E 210	8.8	1.94	7.69	15.53	28.68	35.20					84.75	100.00	5.30	35.20	49.54	15.25
725E 220	7.2	2.20	6.70	14.33	27.33	32.98					87.54	100.00	1.80	32.98	54.56	12.46
725E 230	6.9	2.09	5.79	12.38	23.58	29.29					87.17	100.00	1.19	29.29	57.88	12.83
725E 240	7.0	0.77	2.92	8.77	21.60	27.76					83.51	100.00	0.59	27.76	55.75	16.49
725E 250	7.3	1.86	5.02	11.13	22.17	26.50					87.11	100.00	1.32	26.50	60.60	12.89
725E 255	7.2	2.15	6.29	13.81	26.26	32.77					84.99	100.00	1.42	32.77	52.22	15.01
726E 10	5.8	1.17	3.43	7.76	15.52	19.88					79.90	100.00	0.31	19.88	60.03	20.10
726E 20	7.3	1.45	4.25	9.02	17.98	22.62					82.76	100.00	5.61	22.62	60.14	17.24
726E 30	4.7	2.29	5.98	11.52	21.45	26.46					81.29	100.00	0.41	26.46	54.83	18.71
726E 40	6.5	1.84	5.82	11.75	23.11	28.90					83.57	100.00	0.30	28.90	54.67	16.43
726E 50	5.4	8.07	15.92	21.53	33.31	38.46					75.96	100.00	3.10	38.46	37.50	24.04
726E 55	5.1	2.08	6.66	14.62	26.04	31.48					86.79	100.00	1.10	31.48	55.31	13.21
726E 60	5.1	2.48	7.58	15.77	28.30	33.42					84.04	100.00	1.39	33.42	50.62	15.96



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -																%Sand + Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve intervals) - - - - -																				
		10-20	10-40	10-60	10-140	10-230																
726E 70	4.3	3.82	9.48	16.29	29.04	37.86	87.15	100.00	4.61	37.86	49.29	12.85										
726E 80	5.1	2.45	7.19	12.38	23.52	28.92	91.04	100.00	0.80	28.92	62.12	8.96										
726E 90	4.2	1.84	6.29	11.63	26.30	33.16	90.38	100.00	0.60	33.16	57.22	9.62										
726E 100	3.2	3.93	10.27	17.92	29.90	35.50	89.71	100.00	6.95	35.50	54.21	10.29										
726E 120	4.0	1.98	6.43	11.74	23.30	29.53	89.01	100.00	0.19	29.53	59.48	10.99										
726E 140	3.8	2.56	8.28	15.19	29.26	35.54	91.01	100.00	0.89	35.54	55.47	8.99										
726E 150	4.2	3.09	8.87	16.82	32.28	38.86	90.40	100.00	1.91	38.86	51.54	9.60										
726E 160	3.2	2.98	8.79	12.46	19.21	22.54	91.84	100.00	1.00	22.54	69.30	8.16										
726E 170	2.5	2.05	6.96	14.99	27.57	33.76	89.90	100.00	0.90	33.76	56.14	10.10										
726E 180	3.1	1.63	6.54	14.70	29.48	35.75	88.49	100.00	1.04	35.75	52.74	11.51										
726E 190	5.5	6.86	13.13	20.33	34.51	39.88	87.07	100.00	2.11	39.88	47.19	12.93										
728E 20	5.7	0.16	0.80	10.40	27.73	35.09	96.16	100.00	0.0	35.09	61.08	3.84										
728E 30	8.6	2.88	8.49	8.93	10.12	14.23	72.15	100.00	0.50	14.23	57.92	27.85										
728E 40	5.2	3.08	8.20	20.98	39.97	47.34	83.78	100.00	3.31	47.34	36.44	16.22										
728E 50	8.2	3.66	11.51	17.25	29.12	33.86	89.30	100.00	0.82	33.86	55.44	10.70										
728E 60	3.9	2.76	8.55	17.45	33.49	41.13	91.64	100.00	1.00	41.13	50.51	8.36										
728E 70	5.1	3.95	9.82	18.04	32.53	38.80	90.16	100.00	5.14	38.80	51.36	9.83										
728E 80	4.1	3.23	9.04	18.20	32.57	39.40	88.68	100.00	5.10	39.40	49.29	11.32										
728E 88	3.8	2.36	7.74	16.18	31.76	38.67	90.30	100.00	0.49	38.67	51.63	9.70										
729E 10	5.7	1.89	6.24	14.81	29.90	48.00	91.25	100.00	0.50	48.00	43.26	8.75										
729E 20	5.2	5.37	15.88	23.69	36.20	41.33	84.55	100.00	2.00	41.33	43.22	15.45										
729E 30	6.8	1.19	4.69	18.35	39.80	47.41	95.05	100.00	0.69	47.41	47.64	4.95										





Hole/Depth	CaCO3	-- Cumulative - Percentages --														%Sand	%Silt	%Clay
		-- Sand (Sieve intervals) --																
		10-20	20-40	40-60	60-100	100-200	200-400	400-800	800-1600	1600-3200	3200-6400	6400-12800	12800-25600	25600-51200	51200-102400			
	Equiv.	10-20	20-40	40-60	60-100	100-200	%Sand ↓ Silt	%Sand, Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay						
756E 5	7.3	3.38	9.23	18.50	36.31	41.74	84.62	100.00	1.90	41.74	42.87	15.38						
756E 15	10.5	3.58	11.01	24.33	51.21	56.12	90.01	100.00	1.38	56.12	33.90	9.99						
756E 29	8.0	4.14	11.14	24.39	41.82	47.51	87.56	100.00	2.24	47.51	40.05	12.44						
756E 40	7.1	6.78	11.79	24.14	41.12	46.61	88.65	100.00	1.89	46.61	42.03	11.35						
756E 50	7.6	4.41	12.26	25.27	44.92	50.73	90.45	100.00	4.47	50.73	39.72	9.55						
756E 60	7.2	4.47	11.26	22.87	40.98	46.37	87.03	100.00	4.00	46.37	40.66	12.97						
756E 70	7.7	2.39	11.95	23.65	41.37	47.24	91.41	100.00	1.70	47.24	44.17	8.59						
756E 75	6.3	4.66	12.01	23.54	42.77	46.44	88.27	100.00	5.42	46.44	41.83	11.73						
756E 80	7.6	4.17	11.13	20.69	41.97	46.64	92.48	100.00	2.70	46.64	45.84	7.52						
756E 90	7.2	9.17	14.29	27.50	46.94	52.48	90.50	100.00	4.17	52.48	38.01	9.50						
756E 100	7.3	4.35	11.71	21.95	42.75	50.51	92.13	100.00	3.50	50.51	41.62	7.87						
756E 105	6.4	6.00	15.12	26.47	48.28	52.88	88.48	100.00	3.00	52.88	35.60	11.52						
756E 110	7.0	4.83	12.17	22.00	41.50	46.68	92.44	100.00	5.33	46.68	45.75	7.56						
756E 120	10.1	3.26	3.42	12.32	30.06	35.90	88.28	100.00	2.40	35.90	52.38	11.72						
756E 130	9.0	3.42	8.62	18.63	33.66	38.58	91.80	100.00	2.60	38.58	53.22	8.20						
756E 135	7.8	3.64	9.03	17.18	31.78	36.84	85.66	100.00	4.60	36.84	48.82	14.34						
757E 10	5.9	3.50	10.79	20.68	38.05	44.93	81.47	100.00	0.01	44.93	36.54	18.53						
757E 20	5.1	7.16	16.61	27.23	47.29	54.12	92.20	100.00	2.14	54.12	38.09	7.80						
757E 30	6.5	4.39	11.35	22.73	40.74	47.64	90.61	100.00	5.29	47.64	42.97	9.39						
757E 40	7.0	4.32	11.66	22.47	40.68	47.35	90.69	100.00	1.71	47.35	43.34	9.31						
757E 48	6.9	5.95	12.61	26.14	44.27	50.61	91.79	100.00	10.07	50.61	41.18	8.21						
757E 50	7.7	4.58	12.35	23.30	40.96	47.87	89.64	100.00	4.08	47.87	41.78	10.36						
757E 60	7.2	6.22	14.32	25.05	41.56	47.88	86.43	100.00	1.28	47.88	38.55	13.57						
757E 70	7.2	3.72	10.63	21.39	38.57	45.42	89.29	100.00	2.60	45.42	43.87	10.71						



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand + Silt & Clay 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230									
757E 80	6.3	6.53	13.99	25.23	42.60	48.68	84.89	100.00	4.30			48.68	36.21	15.11	
757E 90	6.5	5.29	12.90	23.43	40.91	47.26	89.34	100.00	3.38			47.26	42.07	10.66	
757E 100	6.9	5.13	12.60	23.37	40.84	47.59	84.68	100.00	3.60			47.59	37.08	15.32	
757E 108	7.2	6.32	14.17	24.82	41.84	48.65	84.93	100.00	5.00			48.65	36.28	15.07	
757E 320	5.4	1.14	3.99	9.01	16.25	19.88	91.44	100.00	0.10			19.88	71.56	8.56	
758E 60	6.5	3.65	8.78	16.38	28.67	34.22	89.22	100.00	11.62			34.22	55.01	10.78	
758E 68	7.0	3.29	9.86	21.48	38.70	45.26	88.61	100.00	0.10			45.26	43.34	11.39	
758E 90	7.3	3.08	9.87	21.34	39.04	45.61	89.63	100.00	1.45			45.61	44.01	10.37	
758E 120	3.5	3.22	9.28	19.02	34.02	40.77	93.03	100.00	1.09			40.77	52.26	6.97	
758E 150	2.3	1.59	7.72	16.53	11.57	36.97	88.67	100.00	0.59			36.97	51.70	11.33	
758E 180	2.6	0.93	6.23	13.55	26.14	33.07	86.59	100.00	0.81			33.07	53.52	13.41	
758E 210	2.9	2.29	7.63	15.83	28.91	35.46	88.23	100.00	1.31			35.46	52.77	11.77	
758E 240	3.0	1.57	6.67	14.88	29.58	36.71	80.38	100.00	0.10			36.71	43.66	19.62	
758E 270	2.7	1.92	6.84	13.42	25.11	31.09	88.93	100.00	0.40			31.09	57.83	11.07	
758E 300	3.0	2.64	7.50	15.88	30.30	36.65	94.61	100.00	2.33			36.65	57.96	5.39	
758E 330	5.2	1.58	4.53	9.49	18.01	22.26	85.15	100.00	0.20			22.26	62.89	14.85	
759E 20	4.1	4.81	10.97	20.23	36.06	41.45	85.95	100.00	4.03			41.45	44.51	14.05	
759E 30	3.9	2.39	7.71	16.34	32.21	38.46	89.25	100.00	0.74			38.46	50.79	10.75	
759E 43	3.8	3.23	8.61	14.49	36.71	46.80	98.20	100.00	1.99			46.80	51.41	1.80	
759E 50	4.4	2.23	7.24	14.58	29.65	36.14	87.30	100.00	1.80			36.14	51.17	12.70	
760E 10	9.7	3.61	9.41	20.03	35.58	39.85	84.51	100.00	4.34			39.85	44.66	15.49	
760E 20	9.6	5.29	12.87	22.68	44.67	49.94	88.40	100.00	2.80			49.94	38.47	11.60	



Hole/Depth	Cac03	Cumulative - Percentages															%Sand, %Coarser than Silt & Clay 10 mesh	%Sand	%Silt	%Clay
		Equiv. - - - Sand (Sieve intervals)																		
		10-20	10-40	10-60	10-140	10-230	%Sand + Silt	%Sand, %Coarser than Silt & Clay 10 mesh												
7602 30	9.8	4.77	11.64	21.66	42.41	47.79	88.87	100.00	2.20	47.79	41.09	11.13								
7602 40	11.3	5.36	13.01	23.51	43.66	50.45	90.47	100.00	4.48	50.45	40.02	9.53								
7602 50	9.5	5.13	12.59	24.89	42.57	49.23	89.66	100.00	3.80	49.23	40.42	10.34								
7602 60	8.6	5.58	13.15	27.78	43.26	50.25	92.58	100.00	2.54	50.25	42.34	7.44								
7602 65	10.3	4.93	12.88	23.15	42.83	48.68	93.90	100.00	3.20	48.68	45.21	6.10								
7602 70	10.5	5.54	13.22	22.07	40.79	48.45	94.25	100.00	6.20	48.45	45.80	5.75								
7602 75	5.5	3.32	8.87	18.53	34.57	40.15	86.47	100.00	1.06	40.15	46.31	13.53								
7602 80	10.7	5.78	13.60	23.55	42.12	49.77	93.42	100.00	3.26	49.77	43.65	6.58								
7602 90	8.2	4.61	13.30	27.95	51.22	59.32	93.75	100.00	1.60	59.32	34.43	6.25								
7602 100	8.1	4.74	12.11	22.52	41.93	47.07	92.27	100.00	2.17	47.07	45.19	7.73								
7602 110	8.3	4.03	12.81	25.15	47.41	54.11	91.54	100.00	1.68	54.11	37.44	8.46								
7602 120	9.9	4.27	14.13	25.97	53.36	64.19	92.47	100.00	1.70	64.19	28.28	7.53								
7612 20	7.8	3.52	9.00	17.32	33.57	40.44	84.73	100.00	7.11	40.44	44.30	15.27								
7612 25	6.6	3.12	7.06	12.84	25.04	31.86	83.54	100.00	10.34	31.86	51.68	16.46								
7612 40	11.0	4.82	11.30	20.72	40.01	47.93	88.35	100.00	4.40	47.93	40.42	11.65								
7612 50	9.9	5.20	13.04	23.98	43.31	51.15	90.06	100.00	4.55	51.15	38.91	9.94								
7612 60	11.2	5.00	12.99	23.96	42.82	51.21	90.40	100.00	2.72	51.21	39.19	9.60								
7612 70	9.8	2.89	7.88	14.42	15.40	15.74	85.69	100.00	1.71	15.74	69.96	14.31								
7612 75	10.4	5.45	14.41	20.62	21.06	21.67	85.38	100.00	0.92	21.67	63.70	14.62								
7612 80	9.1	3.90	10.32	17.46	17.82	18.47	91.88	100.00	1.22	18.47	73.41	8.12								
7612 90	9.8	2.42	6.20	11.14	23.16	30.86	83.10	100.00	0.20	30.86	52.24	16.90								
7612 100	11.9	5.70	12.34	19.67	20.04	20.77	90.91	100.00	12.20	20.77	70.13	9.09								
7612 105	9.3	5.89	13.86	24.24	44.66	51.89	92.40	100.00	1.18	51.89	40.51	7.60								
7612 115	8.8	4.52	12.53	22.76	42.20	48.15	91.60	100.00	1.10	48.15	43.45	8.40								



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															%Sand + Silt	%Sand, %Coarser than Silt & Clay 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230	10-425	10-600	10-840	10-1060	10-1490	10-2000	10-2980	10-4750	10-7500	10-11900					
761E 125	10.1	3.61	10.61	19.90	20.60	20.90	20.90	20.90	20.90	20.90	20.90	20.90	20.90	20.90	20.90	20.90	20.90	73.48	5.62		
761E 135	11.5	3.74	7.48	22.96	27.74	27.97	27.97	27.97	27.97	27.97	27.97	27.97	27.97	27.97	27.97	27.97	27.97	67.32	4.71		
761E 150	10.0	5.06	11.97	20.66	21.48	21.75	21.75	21.75	21.75	21.75	21.75	21.75	21.75	21.75	21.75	21.75	21.75	69.63	8.62		
762E 20	10.2	0.0	0.0	5.68	5.80	6.48	6.48	6.48	6.48	6.48	6.48	6.48	6.48	6.48	6.48	6.48	6.48	75.46	18.06		
762E 240	8.3	1.80	4.50	9.64	24.32	32.63	32.63	32.63	32.63	32.63	32.63	32.63	32.63	32.63	32.63	32.63	32.63	48.05	19.32		
762E 245	9.2	2.58	6.53	13.87	25.58	30.89	30.89	30.89	30.89	30.89	30.89	30.89	30.89	30.89	30.89	30.89	30.89	66.05	3.06		
763E 40	16.6	2.49	8.54	12.41	22.37	27.72	27.72	27.72	27.72	27.72	27.72	27.72	27.72	27.72	27.72	27.72	27.72	65.10	7.18		
763E 50	16.4	2.75	9.82	16.77	36.18	42.89	42.89	42.89	42.89	42.89	42.89	42.89	42.89	42.89	42.89	42.89	42.89	46.76	10.35		
763E 60	9.7	4.92	13.11	24.44	45.20	53.78	53.78	53.78	53.78	53.78	53.78	53.78	53.78	53.78	53.78	53.78	53.78	39.90	6.32		
763E 75	9.6	5.43	15.62	28.43	47.75	55.17	55.17	55.17	55.17	55.17	55.17	55.17	55.17	55.17	55.17	55.17	55.17	35.33	9.49		
763E 80	8.7	4.17	10.78	18.03	40.67	49.28	49.28	49.28	49.28	49.28	49.28	49.28	49.28	49.28	49.28	49.28	49.28	44.87	5.84		
763E 90	9.8	4.38	12.20	21.82	42.28	50.26	50.26	50.26	50.26	50.26	50.26	50.26	50.26	50.26	50.26	50.26	50.26	43.50	6.24		
763E 100	9.2	5.00	13.35	25.92	49.32	58.17	58.17	58.17	58.17	58.17	58.17	58.17	58.17	58.17	58.17	58.17	58.17	36.77	5.06		
763E 110	8.8	4.19	13.68	26.52	51.75	59.58	59.58	59.58	59.58	59.58	59.58	59.58	59.58	59.58	59.58	59.58	59.58	34.49	5.93		
764E 5	8.4	2.98	10.15	19.53	35.04	42.34	42.34	42.34	42.34	42.34	42.34	42.34	42.34	42.34	42.34	42.34	42.34	41.15	16.50		
764E 45	10.8	4.62	12.29	22.19	42.11	49.32	49.32	49.32	49.32	49.32	49.32	49.32	49.32	49.32	49.32	49.32	49.32	37.11	13.57		
764E 55	11.2	4.80	12.58	25.70	41.41	49.36	49.36	49.36	49.36	49.36	49.36	49.36	49.36	49.36	49.36	49.36	49.36	43.61	7.03		
764E 65	10.6	4.95	15.64	32.09	56.55	64.69	64.69	64.69	64.69	64.69	64.69	64.69	64.69	64.69	64.69	64.69	64.69	29.49	5.83		
764E 90	13.0	4.00	8.84	24.86	46.45	54.55	54.55	54.55	54.55	54.55	54.55	54.55	54.55	54.55	54.55	54.55	54.55	37.75	7.70		
764E 98	12.8	4.21	11.06	19.94	39.37	44.23	44.23	44.23	44.23	44.23	44.23	44.23	44.23	44.23	44.23	44.23	44.23	45.69	10.08		
764E 111	13.9	4.66	13.16	25.16	48.61	53.83	53.83	53.83	53.83	53.83	53.83	53.83	53.83	53.83	53.83	53.83	53.83	37.18	8.99		
764E 120	12.5	4.77	13.10	23.93	39.30	43.86	43.86	43.86	43.86	43.86	43.86	43.86	43.86	43.86	43.86	43.86	43.86	46.33	9.82		





Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															
		Sand (Sieve intervals) - - - - -															
		10-20	10-40	10-60	10-140	10-230	%Sand + Silt	%Sand, %Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay					
764E 125	11.9	4.48	12.07	21.43	42.20	47.73	94.05	100.00	2.79	47.73	46.32	5.95					
764E 135	9.3	4.39	11.73	21.40	37.39	44.51	88.79	100.00	7.65	44.51	44.28	11.21					
764E 144	9.7	3.65	10.26	21.42	39.11	43.71	92.17	100.00	8.19	43.71	48.46	7.83					
764E 155	13.7	4.44	10.85	18.54	31.78	38.04	94.30	100.00	1.90	38.04	56.26	5.70					
765E 30	7.9	2.51	8.09	15.61	31.29	36.60	83.50	100.00	0.60	36.60	46.90	16.50					
765E 50	9.1	2.22	5.63	11.29	25.63	32.35	82.43	100.00	2.90	32.35	50.08	17.57					
765E 70	7.0	2.74	5.19	9.09	16.54	19.72	70.64	100.00	1.00	19.72	50.92	29.36					
765E 90	10.6	5.05	13.63	24.28	42.22	51.00	93.46	100.00	2.52	51.00	42.45	6.54					
765E 110	10.4	5.62	14.25	25.77	44.75	53.97	93.64	100.00	4.50	53.97	39.67	6.36					
765E 247	4.6	4.36	12.48	22.28	37.61	46.16	92.51	100.00	2.70	46.16	46.35	7.49					
765E 300	7.9	5.05	12.93	22.57	41.30	48.68	94.08	100.00	2.20	48.68	45.40	5.92					
795E 20	7.3	4.31	11.83	21.50	41.65	49.92	89.22	100.00	2.58	49.92	39.30	10.78					
795E 25	6.1	3.28	10.31	21.71	43.06	50.43	86.86	100.00	2.40	50.43	36.43	13.14					
795E 50	7.7	2.69	9.04	18.04	37.93	47.18	89.90	100.00	1.60	47.18	42.72	10.10					
795E 75	7.6	4.60	11.71	22.95	39.57	46.95	91.33	100.00	5.33	46.95	44.37	8.67					
796E 145	7.0	6.56	14.63	25.66	45.02	52.57	91.84	100.00	4.45	52.57	39.27	8.16					
797E 20	8.6	4.65	11.75	23.63	39.16	45.68	89.61	100.00	7.20	45.68	43.93	10.39					
797E 40	5.9	3.70	10.36	20.80	38.08	45.26	89.99	100.00	2.50	45.26	44.72	10.01					
797E 60	5.7	3.39	9.67	19.10	37.29	44.94	92.34	100.00	0.93	44.94	47.40	7.66					
797E 80	5.7	3.84	11.05	21.00	39.50	46.02	89.89	100.00	1.50	46.02	43.87	10.11					



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															
		Sand (Sieve intervals)										%Sand + Silt	%Sand, Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230											
797E 100	6.1	3.60	9.96	18.73	37.11	43.80	88.11	100.00	3.03	43.80	44.31	11.89					
797E 170	3.9	2.11	6.62	13.42	25.59	33.05	90.77	100.00	2.25	33.05	57.73	9.23					
797E 190	3.7	2.50	7.78	15.23	31.75	36.96	92.49	100.00	0.90	36.96	55.53	7.51					
797E 230	3.6	2.53	8.21	16.37	30.88	41.25	91.00	100.00	0.85	41.25	49.75	9.00					
797E 250	4.4	3.36	6.97	13.23	24.03	31.13	88.66	100.00	3.72	31.13	57.52	11.34					
797E 270	4.4	2.40	7.43	15.07	28.88	35.20	86.86	100.00	1.85	35.20	51.66	13.14					
797E 290	5.2	4.26	8.90	15.15	24.65	29.28	82.43	100.00	10.82	29.28	53.14	17.57					
797E 310	4.8	1.80	6.42	14.02	29.77	35.81	93.51	100.00	0.73	35.81	57.70	6.49					
797E 335	2.3	2.38	7.25	15.81	28.46	36.44	89.36	100.00	1.59	36.44	52.93	10.64					
797E 340	2.8	1.43	4.06	11.39	34.11	42.98	87.51	100.00	0.31	42.98	44.54	12.49					
798E 10	3.6	2.58	7.78	17.87	32.82	38.33	85.41	100.00	0.41	38.33	47.08	14.59					
798E 30	4.3	2.10	6.73	15.54	32.51	37.73	84.09	100.00	2.15	37.73	46.35	15.91					
798E 50	3.2	2.04	6.58	14.23	30.52	36.28	84.38	100.00	0.50	36.28	48.10	15.62					
798E 70	4.1	4.00	9.75	16.67	32.23	38.81	90.80	100.00	2.90	38.81	51.99	9.20					
798E 90	13.2	4.43	10.39	18.61	33.05	40.03	89.86	100.00	3.10	40.03	49.83	10.14					
798E 110	10.6	5.44	12.50	20.11	39.71	47.40	91.18	100.00	4.25	47.40	43.78	8.82					
798E 130	9.3	4.25	7.00	9.84	31.31	38.58	89.82	100.00	6.70	38.58	51.24	10.18					
798E 150	8.9	2.77	7.59	16.15	34.03	39.86	91.33	100.00	2.19	39.86	51.47	8.67					
798E 170	8.7	2.80	6.92	16.85	31.16	35.26	88.95	100.00	2.50	35.26	53.68	11.05					
798E 210	6.1	3.92	11.31	22.24	40.18	47.02	94.84	100.00	1.78	47.02	47.82	5.16					
798E 230	4.6	2.16	4.55	16.93	35.24	43.43	92.94	100.00	3.49	43.43	49.52	7.06					
798E 250	3.4	1.86	4.39	12.12	26.45	31.07	91.15	100.00	1.76	31.07	60.08	8.85					
799E 10	10.2	1.80	4.30	11.54	28.67	35.50	85.22	100.00	0.59	35.50	49.72	14.78					



Hole/Depth	CaCO3 Equiv.	- - - - - Cumulative - Percentages - - - - -																
		Sand (Sieve Intervals)										%Sand + Silt		%Sand, Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230												
799E 20	3.0	1.87	5.04	13.17	28.24	34.56		86.43	100.00	2.55	34.56	51.87	13.57					
799E 30	22.1	3.77	8.34	13.66	27.56	34.80		92.67	100.00	3.10	34.80	57.87	7.33					
799E 40	20.0	1.90	3.97	9.22	20.17	24.31		93.97	100.00	2.30	24.31	69.65	6.03					
799E 50	19.5	5.52	12.38	20.59	36.06	42.53		93.59	100.00	2.39	42.53	51.06	6.41					
799E 100	8.2	3.91	8.97	20.62	45.09	51.71		94.97	100.00	1.43	51.71	43.27	5.03					
799E 180	4.7	2.84	8.13	16.74	29.66	34.19		90.36	100.00	0.79	34.19	56.17	9.64					
799E 185	6.1	5.14	13.27	24.74	38.19	43.21		91.09	100.00	2.09	43.21	47.88	8.91					
801E 10	9.0	2.59	7.57	16.22	32.00	38.61		85.88	100.00	2.19	38.61	47.27	14.12					
801E 20	9.1	2.47	7.47	14.89	28.93	35.29		85.71	100.00	0.70	35.29	50.42	14.29					
801E 30	9.1	4.43	11.41	22.48	45.47	52.21		92.72	100.00	4.39	52.21	40.51	7.28					
801E 40	11.3	2.19	5.01	9.80	21.93	26.36		87.44	100.00	1.50	26.36	61.07	12.56					
801E 80	11.0	2.98	7.43	14.89	27.89	33.80		92.93	100.00	2.30	33.80	59.13	7.07					
801E 85	5.8	3.47	10.25	20.90	38.25	43.67		95.32	100.00	1.40	43.67	51.66	4.63					
801E 90	7.0	3.84	10.07	20.51	39.94	43.83		94.88	100.00	2.00	43.83	51.05	5.12					
801E 190	6.7	4.13	10.91	20.46	36.19	41.81		91.47	100.00	2.30	41.81	49.66	8.53					
801E 195	4.7	5.41	13.48	29.17	48.06	51.60		92.50	100.00	2.90	51.60	40.90	7.50					
801E 200	6.1	3.62	10.17	22.98	40.18	46.37		96.19	100.00	3.23	46.37	49.82	3.81					
802E 10	10.4	2.72	6.86	20.33	40.02	47.88		93.76	100.00	2.10	47.88	45.88	6.24					
802E 15	8.1	2.50	6.99	17.53	38.17	46.98		94.82	100.00	3.02	46.98	47.83	5.18					
802E 22	11.3	1.37	3.76	11.71	34.00	42.06		92.55	100.00	2.20	42.06	50.49	7.45					
802E 25	11.2	1.81	5.35	16.63	36.06	42.96		92.95	100.00	1.09	42.96	49.98	7.05					
802E 65	26.6	8.59	16.37	27.26	44.68	49.78		93.84	100.00	24.30	49.78	44.06	6.16					
802E 67	19.6	4.48	10.86	17.73	36.41	42.61		92.51	100.00	3.90	42.61	49.90	7.49					





Hole/Depth	CAC03	Cumulative - Percentages - - - - -											- - - - -			
	Equiv.	Sand (Sieve intervals)											- - - - -			
		10-20	10-40	10-60	10-140	10-230	%Sand + Silt	%Sand, %Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay				
802E 75	18.2	3.39	7.32	14.05	27.42	32.76	96.98	100.00	7.56	32.76	64.21	3.02				
802E 85	13.7	4.57	10.98	22.87	36.89	40.39	90.24	100.00	5.51	40.39	49.85	9.76				
802E 95	7.0	3.36	9.08	19.05	35.14	39.19	95.06	100.00	3.50	39.19	55.87	4.94				
802E 110	7.8	4.33	13.86	33.81	49.60	51.93	92.89	100.00	3.32	51.93	40.96	7.11				
802E 115	9.7	3.56	10.14	19.99	39.69	44.82	97.33	100.00	5.48	44.82	52.52	2.67				
802E 140	2.9	2.10	6.88	16.96	31.64	37.62	97.18	100.00	4.28	37.62	59.56	2.82				
802E 175	6.9	2.82	7.47	11.81	19.37	22.76	97.24	100.00	1.80	22.76	74.47	2.76				
802E 185	7.0	2.84	7.72	13.02	21.87	25.65	96.73	100.00	2.10	25.65	71.08	3.27				
802E 195	3.9	0.80	2.39	12.29	33.64	40.16	92.59	100.00	0.10	40.16	52.43	7.41				
804E 10	4.0	2.39	4.48	6.95	11.42	14.60	82.15	100.00	0.70	14.60	67.55	17.85				
804E 20	9.8	0.23	0.46	0.46	0.75	0.85	98.85	100.00	0.0	0.85	98.00	1.15				
804E 25	9.1	0.04	0.33	0.58	0.82	1.07	87.52	100.00	0.08	1.07	86.46	12.48				
804E 30	10.9	0.22	0.75	1.37	2.41	3.03	90.80	100.00	0.08	3.03	87.77	9.20				
804E 55	17.1	4.80	10.70	20.34	37.05	42.09	91.40	100.00	3.40	42.09	49.30	8.60				
804E 60	17.4	3.97	9.59	16.94	34.04	40.35	93.24	100.00	2.77	40.35	52.89	6.76				
804E 65	15.6	3.62	7.73	15.86	32.00	37.56	91.79	100.00	4.61	37.56	54.22	8.21				
804E 70	14.5	3.57	8.44	17.65	34.82	38.97	90.06	100.00	3.50	38.97	51.09	9.94				
804E 75	14.9	2.84	7.81	19.11	33.76	38.29	91.49	100.00	1.76	38.29	53.20	8.51				
804E 80	15.0	3.93	9.36	18.28	34.69	40.50	91.12	100.00	4.10	40.50	50.62	8.88				
804E 95	15.0	4.62	10.80	21.56	42.11	48.28	92.62	100.00	2.50	48.28	44.34	7.38				
804E 100	13.7	3.40	8.72	17.00	35.77	39.78	91.40	100.00	5.60	39.78	51.62	8.60				
804E 105	11.3	3.22	9.17	11.08	32.30	38.00	91.18	100.00	2.75	38.00	53.19	8.82				
804E 108	9.7	4.60	11.70	19.41	38.66	47.31	88.68	100.00	3.70	47.31	41.37	11.32				



Hole/Depth	CaCO <sub>3</sub> Equiv.	Cumulative - Percentages - - - - -										
		Sand (Sieve intervals) - - - - -					%Sand + Silt		%Sand, %Silt & Clay			
		10-20	20-40	40-60	60-100	100-200	%Sand	%Silt	%Sand	%Silt	%Sand	%Clay
804E 110	10.1	5.64	11.60	20.98	37.41	44.72	94.38	100.00	2.65	44.72	49.66	5.62
804E 115	9.4	4.85	11.57	20.18	37.33	43.93	98.34	100.00	2.99	43.93	54.41	1.66
804E 120	9.8	4.45	11.29	20.45	38.89	45.71	95.33	100.00	2.29	45.71	49.63	4.67
804E 125	5.4	4.20	13.02	21.50	35.36	41.87	95.64	100.00	3.08	41.87	53.76	4.36
804E 130	5.2	3.33	6.37	12.37	23.51	29.22	89.25	100.00	3.22	29.22	60.03	10.75
804E 185	2.3	0.60	1.50	1.97	2.53	4.26	98.96	100.00	0.0	4.26	94.69	1.04
804E 190	5.1	0.08	0.19	0.54	10.62	22.09	93.00	100.00	0.0	22.09	70.91	7.00
804E 195	4.3	0.45	0.87	1.22	1.82	2.73	92.51	100.00	0.10	2.73	89.78	7.49
804E 200	4.3	0.62	1.13	2.28	4.11	4.84	93.64	100.00	0.12	4.84	88.80	6.36
804E 205	6.1	2.93	7.49	15.00	26.92	31.74	92.02	100.00	1.18	31.74	60.28	7.98
804E 210	6.9	4.46	6.48	13.93	26.18	31.54	91.76	100.00	3.80	31.54	60.22	8.24
804E 215	5.1	2.37	6.81	9.81	20.71	25.21	89.33	100.00	1.40	25.21	64.12	10.67
806E 5	10.4	2.50	7.27	13.27	28.90	35.79	89.49	100.00	1.58	35.79	53.70	10.51
806E 10	10.4	2.35	7.14	13.43	30.46	36.01	89.70	100.00	1.02	36.01	53.69	10.30
806E 15	8.0	2.17	5.60	10.86	22.04	26.47	83.80	100.00	0.20	26.47	57.33	16.20
806E 20	14.1	4.06	10.44	17.57	37.15	44.68	92.80	100.00	3.49	44.68	48.12	7.20
806E 30	15.6	3.92	10.11	18.48	37.74	43.83	96.10	100.00	3.08	43.83	52.27	3.90
806E 35	16.1	3.99	10.02	17.46	36.02	42.69	98.25	100.00	2.12	42.69	55.56	1.75
806E 40	16.6	4.24	10.11	17.39	35.62	42.72	92.75	100.00	9.50	42.72	50.04	7.25
806E 50	16.0	3.88	10.29	17.04	35.14	42.80	96.10	100.00	3.10	42.80	53.30	3.90
806E 55	15.6	3.94	9.64	16.52	34.61	41.04	96.09	100.00	5.20	41.04	55.05	3.91
806E 65	8.4	4.18	11.74	20.66	40.63	47.93	95.04	100.00	1.50	47.93	47.12	4.96
806E 70	7.9	4.36	12.22	22.07	43.29	49.86	97.13	100.00	4.60	49.86	47.27	2.87
806E 75	8.7	3.89	11.26	23.17	43.12	49.81	87.63	100.00	3.19	49.81	37.82	12.37



Hole/Depth	CaCO3 Equiv.	-- -- -- -- -- Cumulative - Percentages -- -- -- -- --										%Sand, %Coarser Silt than 10 mesh	%Sand	%Silt	%Clay
		-- -- -- -- -- Sand (Sieve intervals) -- -- -- -- --													
		10-20	10-40	10-60	10-100	10-200	10-230	%Sand + Silt & Clay							
806E 85	7.5	4.34	11.90	23.03	41.59	48.63	90.02	100.00	4.40	48.63	41.40	9.98			
806E 90	7.3	4.35	12.08	22.74	42.15	48.78	71.04	100.00	6.21	48.78	22.26	28.96			
806E 95	7.9	4.22	12.71	23.35	41.52	49.05	92.92	100.00	1.52	49.05	43.87	7.08			
806E 100	8.0	4.02	12.14	23.76	45.40	51.41	93.09	100.00	1.10	51.41	41.69	6.91			
806E 105	9.4	4.60	11.81	23.27	42.13	48.81	84.41	100.00	6.99	48.81	35.59	15.59			
806E 115	9.8	3.74	10.81	19.48	35.57	42.92	90.90	100.00	1.78	42.92	47.98	9.10			
806E 120	9.8	4.38	12.12	22.33	39.54	46.81	94.18	100.00	2.71	46.81	47.37	5.82			
806E 125	8.6	4.65	11.71	21.24	37.28	44.33	88.07	100.00	2.39	44.33	43.74	11.93			
807E 20	7.1	2.40	7.34	16.52	34.00	38.15	82.20	100.00	0.80	38.15	44.05	17.80			
807E 30	5.9	3.57	8.14	16.52	29.24	33.27	82.11	100.00	0.90	33.27	48.84	17.89			
807E 40	5.7	5.58	10.49	18.55	32.60	36.82	87.34	100.00	2.09	36.82	50.52	12.66			
807E 50	8.4	2.14	5.38	13.43	29.67	34.83	90.23	100.00	2.10	34.83	55.40	9.77			
807E 60	8.3	1.82	4.78	10.71	21.93	25.49	83.93	100.00	0.25	25.49	58.44	16.07			
807E 70	8.7	1.70	4.89	11.42	23.45	26.66	84.59	100.00	0.26	26.66	57.93	15.41			
807E 80	6.5	3.12	7.19	15.40	31.71	37.24	90.55	100.00	7.81	37.24	53.31	9.45			
807E 90	8.0	7.01	12.37	20.45	37.21	41.32	91.58	100.00	12.84	41.32	50.25	8.42			
807E 125	9.3	3.70	8.76	17.49	35.61	38.29	95.00	100.00	3.80	38.29	56.71	5.00			
807E 135	6.2	4.26	11.41	21.74	40.95	45.27	93.72	100.00	5.63	45.27	48.45	6.28			
807E 145	9.4	6.39	13.99	25.28	44.58	48.58	92.38	100.00	5.50	48.58	43.80	7.62			
807E 155	6.9	4.23	9.86	21.56	40.33	44.08	91.15	100.00	3.66	44.08	47.07	8.85			
807E 166	8.0	5.51	13.71	30.28	52.60	56.66	97.56	100.00	1.90	56.66	40.90	2.44			
807E 185	6.6	3.58	7.66	17.47	31.46	36.19	87.30	100.00	2.60	36.19	51.11	12.70			
807E 200	5.9	3.87	10.29	23.02	40.85	46.98	90.91	100.00	9.73	46.98	43.93	9.09			
807E 206	8.3	4.36	8.32	21.76	30.71	33.03	86.22	100.00	8.08	33.03	53.18	13.78			



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand + Silt	%Sand, Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		10-20	20-40	40-60	60-100	100-200	200-400	400-800	800-1600	1600-3200	3200-6400						
807E 230	6.2	7.59	15.32	28.46	41.09	44.57	87.85	100.00	1.78	44.57	43.28	12.15					
807E 265	7.6	3.55	8.72	20.00	29.96	32.59	88.16	100.00	2.42	32.59	55.58	11.84					
807E 270	7.3	4.31	10.43	23.70	36.98	40.74	91.54	100.00	1.98	40.74	50.80	8.46					
807E 280	7.5	3.89	9.47	21.63	32.27	34.64	89.91	100.00	1.69	34.64	55.28	10.09					
807E 290	5.4	2.26	5.82	14.26	22.65	24.61	90.83	100.00	1.30	24.61	66.22	9.17					
807E 300	5.8	1.21	3.22	8.87	17.84	20.21	83.74	100.00	0.45	20.21	63.54	16.26					
807E 310	6.1	3.25	7.02	17.36	26.68	28.34	85.24	100.00	0.79	28.34	56.90	14.76					
308E 5	5.2	1.06	3.99	9.47	18.50	20.89	75.06	100.00	0.20	20.89	54.16	24.94					
808E 25	4.6	2.27	5.22	13.03	26.59	30.70	84.55	100.00	0.61	30.70	53.84	15.45					
808E 35	6.1	2.85	8.46	21.47	32.96	36.16	82.12	100.00	1.18	36.16	45.96	17.88					
808E 45	7.7	1.81	5.96	16.23	29.88	34.27	84.41	100.00	1.28	34.27	50.13	15.59					
808E 55	9.5	2.63	6.69	17.18	34.78	39.88	85.39	100.00	2.61	39.88	45.51	14.61					
808E 65	5.0	1.81	6.55	18.43	31.74	36.20	91.62	100.00	1.09	36.20	55.42	8.38					
808E 75	5.7	2.31	6.90	19.72	31.13	33.50	89.37	100.00	2.38	33.50	55.87	10.63					
808E 85	5.4	1.92	5.14	16.34	30.25	34.07	88.13	100.00	1.08	34.07	54.06	11.87					
808E 95	5.1	1.71	6.07	17.58	29.36	32.97	90.22	100.00	0.93	32.97	57.25	9.78					
808E 105	1.8	1.33	5.14	48.60	63.70	65.86	87.82	100.00	0.20	65.86	21.96	12.18					
808E 125	5.7	1.74	5.88	19.38	31.88	35.66	88.80	100.00	0.82	35.66	53.14	11.20					
808E 134	6.6	1.99	4.55	14.66	29.74	33.35	88.45	100.00	4.70	33.35	55.11	11.55					
808E 185	1.2	1.91	6.98	14.97	26.68	28.98	89.27	100.00	1.35	28.98	60.29	10.73					
808E 195	2.8	2.90	7.39	24.44	45.33	51.93	89.84	100.00	4.45	51.93	37.91	10.16					
808E 206	3.9	1.23	2.63	5.63	11.21	13.64	87.19	100.00	1.59	13.64	73.55	12.81					
808E 215	7.3	0.67	2.22	11.05	20.36	21.75	92.83	100.00	0.0	21.75	71.07	7.17					
808E 320	7.5	3.61	9.18	20.79	32.10	34.93	93.42	100.00	1.40	34.93	58.48	6.58					





Hole/Depth	CaCO <sub>3</sub> Equiv.	Cumulative - Percentages - - - - -										%Sand ↑ Silt	%Sand, %Coarser than Silt & Clay 10 mesh	%Sand	%Silt	%Clay
		10-20	10-40	10-60	10-140	10-230										
808E 325	7.6	4.82	11.41	23.37	36.36	40.78						91.87	100.00	20.78	51.09	8.13
808E 330	7.9	4.21	10.58	21.20	32.21	36.06						90.49	100.00	36.06	54.44	9.51
808E 335	6.8	3.52	8.85	19.02	28.87	35.31						90.60	100.00	35.31	55.29	9.40
808E 340	5.9	2.24	5.04	12.73	20.96	24.27						90.46	100.00	24.27	66.19	9.54
809E 15	2.3	1.77	5.62	15.14	26.53	29.75						84.98	100.00	29.75	55.23	15.02
809E 25	2.5	2.42	7.17	17.07	26.37	28.75						83.76	100.00	28.75	55.01	16.24
809E 35	3.7	2.19	6.97	17.39	30.60	35.71						90.92	100.00	35.71	55.20	9.08
809E 45	2.3	6.12	14.43	27.26	40.08	44.91						86.21	100.00	44.91	41.30	13.79
809E 55	3.4	4.58	10.68	24.41	36.79	40.87						90.29	100.00	40.87	49.42	9.71
809E 65	2.9	2.89	8.54	20.73	32.26	39.03						90.96	100.00	39.03	51.93	9.04
809E 75	3.5	2.65	8.24	21.56	34.19	39.25						93.05	100.00	39.25	53.81	6.95
809E 85	2.2	2.15	6.95	17.81	29.44	33.70						89.45	100.00	33.70	55.75	10.55
809E 95	2.3	3.14	8.36	20.69	33.33	38.38						92.15	100.00	38.38	53.77	7.85
809E 105	1.9	3.18	8.55	20.62	37.44	43.66						93.25	100.00	43.66	49.59	6.75
809E 115	1.3	4.21	11.99	23.24	34.82	38.24						88.99	100.00	38.24	50.75	11.01
828E 10	5.9	2.55	9.58	22.46	39.30	44.40						85.83	100.00	44.40	41.43	14.17
828E 20	6.8	2.13	9.26	24.22	40.78	45.63						92.81	100.00	45.63	47.18	7.19
828E 30	6.3	3.71	10.71	25.77	42.14	46.72						91.93	100.00	46.72	45.20	8.07
828E 40	5.9	3.74	11.02	25.92	38.94	43.97						89.29	100.00	43.97	45.32	10.71
828E 50	7.1	3.55	10.82	24.85	41.93	47.33						89.97	100.00	47.33	42.64	10.03
828E 60	8.1	3.24	9.48	22.88	38.94	44.31						95.33	100.00	44.31	51.02	4.67
828E 70	8.6	3.36	9.77	22.89	38.99	43.71						93.12	100.00	43.71	49.40	6.88



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -										%Sand + Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve intervals) - - - - -														
		10-20	10-40	10-60	10-140	10-230										
829E 5	1.0	2.24	6.90	19.64	34.69	39.15	81.40	100.00	0.45	39.15	42.25	18.60				
829E 10	8.6	1.43	6.84	25.24	52.96	59.33	90.97	100.00	0.30	59.33	31.64	9.03				
829E 15	9.2	2.38	6.74	14.78	34.25	40.61	99.45	100.00	0.90	40.61	58.84	0.55				
829E 20	9.3	2.17	6.66	16.11	33.19	39.27	84.48	100.00	0.50	39.27	45.21	15.52				
829E 25	8.6	5.54	11.51	21.47	38.62	44.18	85.83	100.00	14.48	44.18	41.64	14.17				
829E 30	9.1	2.44	6.77	16.50	36.59	42.26	86.69	100.00	2.50	42.26	44.43	13.31				
829E 35	9.0	2.31	6.67	15.56	35.53	42.95	86.82	100.00	1.35	42.95	43.87	13.18				
829E 40	8.9	2.21	6.12	15.55	29.60	35.99	82.81	100.00	1.02	35.99	46.82	17.19				
829E 45	9.0	1.84	5.32	14.56	28.84	34.58	83.87	100.00	0.50	34.58	49.29	16.13				
829E 50	9.1	2.11	6.24	14.67	30.43	36.36	83.30	100.00	0.56	36.36	46.94	16.70				
829E 55	9.2	1.95	6.02	15.44	27.60	32.62	81.07	100.00	1.40	32.62	48.45	18.93				
829E 60	8.1	1.60	4.93	11.09	22.71	26.70	78.04	100.00	1.12	26.70	51.35	21.96				
829E 65	20.7	3.38	9.51	18.92	29.06	32.30	91.69	100.00	1.70	32.30	59.39	8.31				
829E 70	17.3	3.47	9.06	19.38	33.12	37.24	90.69	100.00	1.20	37.24	53.45	9.31				
829E 75	11.6	2.45	6.40	14.79	23.32	26.44	85.54	100.00	1.39	26.44	59.10	14.46				
829E 80	12.0	3.78	8.78	17.19	28.66	33.05	84.69	100.00	3.38	33.05	51.64	15.31				
829E 85	12.1	2.28	5.62	11.83	19.14	21.76	83.05	100.00	0.81	21.76	61.29	16.95				
829E 90	7.7	5.70	14.70	32.32	49.70	54.93	91.99	100.00	6.99	54.93	37.06	8.01				
829E 95	13.2	4.53	11.62	54.09	31.86	34.75	85.68	100.00	3.70	34.75	50.92	14.32				
830E 5	7.2	2.04	5.43	10.80	28.26	38.61	90.60	100.00	0.70	38.61	52.00	9.40				
830E 10	5.5	1.50	4.42	9.56	19.84	24.70	84.64	100.00	0.20	24.70	59.93	15.36				
830E 45	16.3	4.08	9.96	18.61	36.20	44.09	93.63	100.00	4.20	44.09	49.54	6.37				
830E 65	16.4	4.85	12.74	23.47	43.33	49.94	94.09	100.00	3.71	49.94	44.15	5.91				
830E 70	9.2	5.53	13.33	23.36	43.66	48.86	93.32	100.00	3.70	49.86	43.46	6.68				



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															
		Sand (Sieve Intervals) - - - - -															
		10-20	20-40	40-60	60-100	10-140	10-230	%Sand + Silt	%Sand, Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay				
8302 75	8.9	5.06	12.95	24.68	44.12	51.04	95.08	100.00	10.26	51.04	44.04	4.92					
8302 80	8.4	4.28	11.76	22.81	44.46	51.68	93.44	100.00	2.15	51.68	41.76	6.56					
8302 85	8.3	5.22	12.73	24.48	45.32	52.29	93.24	100.00	3.50	52.29	40.96	6.76					
8302 90	9.0	4.56	12.44	24.79	46.13	53.36	94.75	100.00	3.09	53.36	41.39	5.25					
8302 95	7.9	4.58	12.31	24.74	45.45	52.61	93.54	100.00	1.88	52.61	40.93	6.46					
8302 120	8.6	5.09	13.56	27.67	49.91	57.19	95.09	100.00	3.15	57.19	37.90	4.91					
8302 153	9.9	3.76	9.97	18.09	35.35	42.55	93.69	100.00	5.50	42.55	51.15	6.31					
8302 164	7.7	3.41	8.68	17.61	33.37	39.54	89.84	100.00	9.28	39.54	50.29	10.16					
8312 10	7.3	1.39	4.80	11.22	22.31	26.98	80.99	100.00	0.30	26.98	54.00	19.01					
8312 40	12.0	4.46	10.62	19.47	37.40	44.41	92.65	100.00	10.65	44.41	48.24	7.35					
8312 60	10.8	3.73	9.81	19.61	35.36	42.29	94.63	100.00	2.40	42.29	52.34	5.37					
8312 80	10.9	4.75	11.59	21.99	40.65	48.20	94.32	100.00	1.71	48.20	46.12	5.68					
8312 100	8.0	4.65	11.75	24.29	42.32	49.59	94.39	100.00	7.80	49.59	44.81	5.61					
8312 140	6.9	4.34	11.27	22.62	38.76	46.00	91.80	100.00	1.48	46.00	45.80	8.20					
8312 160	5.7	4.44	10.94	20.32	36.42	42.90	94.80	100.00	2.32	42.90	51.90	5.20					
8312 180	6.4	5.92	13.82	23.60	39.05	45.30	90.47	100.00	4.30	45.30	45.17	9.53					
8312 200	7.3	4.89	12.00	21.68	37.50	44.24	90.64	100.00	3.18	44.24	46.40	9.36					
8312 220	7.0	3.85	10.23	18.97	34.73	41.72	93.01	100.00	3.00	41.72	51.29	6.99					
8312 280	4.7	5.01	12.24	23.41	38.29	43.76	92.62	100.00	6.10	43.76	48.86	7.38					
8322 10	7.3	4.78	9.55	20.21	33.20	38.14	88.85	100.00	6.18	38.14	50.72	11.15					
8322 20	8.7	2.27	6.31	15.00	29.01	34.76	86.20	100.00	0.85	34.76	51.44	13.80					
8322 30	8.3	2.29	6.49	18.64	30.52	35.22	87.40	100.00	0.90	35.22	52.18	12.60					
8322 40	8.0	2.60	6.60	13.57	28.45	34.38	87.38	100.00	1.69	34.38	53.00	12.62					





Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															%Sand + Silt & Clay 10 mesh	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve intervals) - - - - -																			
		10-20	10-40	10-60	10-100	10-200	10-400	10-600	10-800	10-1000	10-1200	10-1400	10-1600	10-1800	10-2000	10-2200	10-2400	10-2600	10-2800	10-3000	
832E 50	6.4	1.56	5.14	11.81	23.25	28.79	85.54	100.00	0.51	28.79	56.75	14.46									
832E 60	7.5	2.79	8.02	16.55	30.91	43.22	87.37	100.00	1.95	43.22	44.15	12.63									
832E 70	10.5	3.22	7.86	15.57	30.65	38.30	85.79	100.00	4.30	38.30	47.49	14.21									
832E 80	9.1	1.73	5.36	9.91	18.05	25.50	80.17	100.00	0.69	25.50	54.67	19.83									
832E 90	10.2	1.40	4.11	11.00	21.77	26.92	87.95	100.00	0.99	26.92	61.03	12.05									
832E 100	17.0	0.75	2.46	11.89	27.46	34.43	86.74	100.00	0.30	34.43	52.31	13.26									
832E 110	8.5	1.20	3.23	7.91	15.42	21.14	85.11	100.00	0.70	21.14	63.98	14.89									
832E 120	8.8	0.97	2.82	10.19	22.33	29.10	87.45	100.00	0.20	29.10	58.35	12.55									
832E 130	21.2	3.08	7.92	16.89	31.06	36.66	89.60	100.00	3.14	36.66	52.94	10.40									
832E 140	15.8	3.73	9.25	18.71	33.74	40.38	91.31	100.00	1.90	40.38	50.92	8.69									
832E 150	7.2	2.84	7.75	11.20	16.30	20.10	85.36	100.00	0.51	20.10	65.26	14.64									
832E 160	7.2	2.90	8.87	19.09	35.45	41.57	88.60	100.00	1.50	41.57	47.03	11.40									
832E 170	6.1	2.39	6.81	16.21	28.24	32.44	81.59	100.00	0.58	32.44	49.15	18.41									
832E 180	6.6	1.55	4.93	16.62	32.11	37.88	87.25	100.00	0.49	37.88	49.37	12.75									
832E 190	3.8	5.17	14.54	25.87	41.74	48.69	90.10	100.00	1.99	48.69	41.40	9.90									
832E 198	3.7	1.90	5.04	10.16	17.57	20.86	79.93	100.00	1.24	20.86	59.07	20.07									
833E 10	5.0	3.57	7.70	15.36	23.97	27.55	85.14	100.00	1.40	27.55	57.59	14.86									
833E 20	5.5	2.19	5.17	13.13	23.49	26.44	89.99	100.00	3.50	26.44	63.55	10.01									
833E 30	5.7	2.11	6.14	14.96	27.38	31.24	88.11	100.00	2.10	31.24	56.87	11.89									
833E 40	7.5	2.14	6.17	14.95	26.24	29.29	86.87	100.00	1.10	29.29	57.57	13.13									
833E 50	5.9	5.21	9.94	19.85	32.68	36.73	88.84	100.00	1.50	36.73	52.11	11.16									
833E 60	6.2	4.26	9.07	17.28	28.59	32.50	87.39	100.00	4.10	32.50	54.89	12.61									
833E 70	7.2	1.17	3.73	9.32	18.60	22.29	81.95	100.00	0.40	22.29	59.66	18.05									
833E 80	5.1	4.25	9.28	18.24	32.44	37.35	91.96	100.00	5.85	37.35	54.62	8.04									



Hole/Depth	CaCO3	Cumulative - Percentages - - - - -										%Sand, Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
Equiv.		10-20	10-40	10-60	10-140	10-230	%Sand + Silt									
833E 90	5.2	2.58	7.41	16.54	31.52	36.21	92.05	100.00	1.44	36.21	55.83	7.95				
833E 100	4.6	3.38	9.14	19.31	36.20	41.39	93.46	100.00	2.80	41.39	52.07	6.54				
833E 115	13.0	3.98	9.85	19.62	33.81	38.56	92.39	100.00	1.22	38.56	53.83	7.61				
833E 125	14.3	3.36	8.30	15.80	27.47	31.89	91.48	100.00	3.62	31.89	59.59	8.52				
833E 135	10.9	3.30	7.22	13.95	23.97	27.26	87.93	100.00	1.70	27.26	60.67	12.07				
833E 200	3.0	2.97	7.42	16.01	30.74	37.06	91.53	100.00	3.00	37.06	54.47	8.47				
833E 210	2.1	2.67	8.33	18.20	32.71	38.93	92.68	100.00	0.70	38.93	53.75	7.32				
833E 220	1.9	2.25	7.10	16.47	30.60	36.56	91.97	100.00	0.60	36.56	55.41	8.03				
833E 230	2.9	2.66	6.69	15.04	28.73	33.90	89.37	100.00	1.28	33.90	55.47	10.63				
833E 240	4.6	1.91	5.42	13.26	24.78	29.57	86.64	100.00	0.40	29.57	57.07	13.36				
833E 250	4.4	3.34	9.06	15.90	28.01	32.60	90.24	100.00	0.51	32.60	57.64	9.76				
933E 260	4.4	1.85	6.25	15.50	27.53	32.19	88.38	100.00	0.20	32.19	56.19	11.62				
833E 270	3.5	2.15	6.04	16.36	27.99	33.13	91.39	100.00	0.35	33.13	58.26	8.61				
833E 280	2.5	0.19	0.86	1.24	6.39	9.43	90.95	100.00	0.0	9.43	81.52	9.05				
834E 5	1.4	2.26	7.84	17.60	32.45	36.89	84.86	100.00	1.19	36.89	47.97	15.14				
834E 15	4.4	1.95	6.54	15.80	30.16	35.39	86.27	100.00	0.20	35.39	50.88	13.73				
834E 25	1.4	2.99	8.27	18.96	33.41	39.33	87.51	100.00	1.20	39.33	48.18	12.49				
834E 35	5.7	4.12	12.05	27.22	46.16	51.44	92.72	100.00	2.20	51.44	41.28	7.28				
834E 45	8.7	4.74	12.20	25.03	42.84	48.54	92.53	100.00	1.70	48.54	43.99	7.47				
834E 55	7.9	3.50	9.53	22.10	36.81	41.78	91.24	100.00	1.92	41.78	49.46	8.76				
834E 65	7.0	5.06	13.03	26.46	40.63	45.24	91.82	100.00	3.28	45.24	46.57	8.18				
834E 85	9.0	4.45	11.80	25.25	38.78	43.08	87.84	100.00	1.23	43.08	44.76	12.16				
834E 95	6.1	3.73	9.29	20.23	33.62	38.41	90.04	100.00	1.98	38.41	51.62	9.96				
834E 160	4.0	3.70	9.88	22.50	38.00	42.55	90.70	100.00	1.65	42.55	48.15	9.30				



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															%Sand, %Silt & Clay	%Sand + Silt than 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve intervals) - - - - -																			
		10-20	10-40	10-60	10-140	10-230															
834E 170	4.3	3.67	9.64	22.16	37.45	42.00						90.06	100.00	42.00	48.06	9.94					
834E 180	3.5	2.38	7.12	19.07	32.02	36.60						89.26	100.00	36.60	52.66	10.74					
834E 190	3.7	2.88	8.77	21.95	36.86	41.59						88.68	100.00	41.59	47.09	11.32					
835E 10	4.0	2.08	6.35	15.41	28.77	34.36						79.96	100.00	34.36	45.60	20.04					
835E 30	3.7	2.12	6.59	15.96	29.02	34.99						82.71	100.00	34.99	47.72	17.29					
835E 40	3.7	2.12	6.94	16.70	29.19	35.54						83.39	100.00	35.54	47.85	16.61					
835E 50	3.8	1.36	5.12	13.35	25.97	31.54						83.13	100.00	31.54	51.59	16.87					
835E 60	3.9	2.01	6.18	13.71	24.76	30.14						83.57	100.00	30.14	53.42	16.43					
835E 70	3.9	1.88	6.33	15.12	28.98	35.58						88.54	100.00	35.58	52.96	11.46					
835E 80	8.1	2.42	7.29	17.27	29.54	37.72						86.51	100.00	37.72	48.79	13.49					
835E 90	6.1	2.24	6.36	13.92	27.60	33.26						87.69	100.00	33.26	54.43	12.31					
835E 100	6.3	4.48	10.75	16.48	26.09	30.59						93.46	100.00	30.59	62.88	6.54					
835E 150	4.4	3.11	9.76	24.54	49.32	59.92						87.54	100.00	59.92	27.62	12.46					
835E 160	3.5	3.03	9.45	17.82	34.15	41.83						90.83	100.00	41.83	49.00	9.17					
835E 170	2.5	2.27	7.41	15.56	28.74	35.57						91.10	100.00	35.57	55.53	8.90					
835E 180	1.2	2.22	7.71	15.95	30.49	37.48						92.35	100.00	37.48	54.87	7.65					
836E 30	7.3	1.79	5.63	12.79	26.83	33.01						84.11	100.00	33.01	51.10	15.89					
836E 60	7.4	2.44	6.44	15.11	29.95	36.15						87.00	100.00	36.15	50.86	13.00					
836E 100	7.4	3.30	9.38	20.32	34.81	41.00						88.69	100.00	41.00	47.69	11.31					
836E 140	6.4	0.97	4.54	13.20	28.04	34.13						84.35	100.00	34.13	50.22	15.65					
836E 200	2.6	1.53	6.07	15.68	29.11	35.68						90.43	100.00	35.68	54.75	9.57					



Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -															%Sand + Silt	%Sand, %Silt & Clay 10 mesh	%Sand	%Silt	%Clay
		Sand (Sieve Intervals) - - - - -																			
		10-20	20-40	40-60	60-100	100-200	200-400	400-800	800-1600	1600-3200	3200-6400	6400-12800	12800-25600	25600-51200	51200-102400	102400-204800					
952E 10	6.8	2.21	6.36	12.61	24.74	30.82	84.23	100.00	1.15	30.82	53.42	15.77									
952E 20	7.6	3.09	7.63	14.69	28.00	33.55	85.29	100.00	6.56	33.55	51.73	14.71									
952E 24	7.1	2.20	5.67	11.27	22.53	28.86	80.19	100.00	3.12	28.86	51.33	19.81									
952E 30	6.5	2.80	7.29	13.14	23.48	27.22	82.22	100.00	1.10	27.22	55.00	17.78									
952E 40	4.9	4.74	11.01	18.94	32.05	37.39	85.78	100.00	10.80	37.39	48.39	14.22									
952E 65	5.1	3.02	8.65	17.48	31.88	38.41	89.23	100.00	2.83	38.41	50.82	10.77									
952E 70	6.2	3.27	9.22	17.05	33.31	39.29	88.43	100.00	1.01	39.29	49.14	11.57									
952E 75	5.7	4.02	9.81	16.90	32.18	38.73	86.41	100.00	8.50	38.73	47.68	13.59									
952E 80	6.0	4.67	11.31	18.92	33.50	39.60	89.63	100.00	12.40	39.60	50.02	10.37									
952E 95	7.3	3.85	11.17	20.73	38.51	45.55	89.85	100.00	4.40	45.55	44.30	10.15									
952E 105	6.8	3.23	9.85	18.37	35.65	43.44	87.37	100.00	0.80	43.44	43.93	12.63									
952E 110	6.8	3.59	9.56	18.26	33.62	39.34	86.97	100.00	8.98	39.34	47.62	13.03									
953E 6	3.9	0.60	6.56	14.84	26.83	31.75	79.46	100.00	0.20	31.75	47.72	20.54									
953E 60	4.4	3.20	8.55	16.65	29.97	37.15	81.90	100.00	4.30	37.15	44.74	18.10									
953E 70	2.7	2.52	7.84	16.22	30.90	37.98	81.01	100.00	0.50	37.98	43.02	18.99									
953E 80	2.1	1.99	7.05	15.91	30.01	36.83	81.97	100.00	0.91	36.83	45.14	18.03									
953E 90	3.2	1.16	4.61	11.78	27.41	34.51	88.85	100.00	1.31	34.51	54.34	11.15									
953E 100	3.0	1.96	7.56	17.12	31.47	37.50	87.65	100.00	7.75	37.50	50.14	12.35									
953E 105	6.2	3.72	9.64	20.07	36.35	42.37	86.22	100.00	5.88	42.37	43.85	13.78									
953E 110	2.7	1.51	6.49	14.02	27.38	34.32	85.42	100.00	0.20	34.32	51.10	14.58									
953E 120	2.5	1.12	4.77	13.25	26.78	33.54	87.56	100.00	0.30	33.54	54.02	12.44									
953E 130	2.5	1.18	5.48	13.73	27.37	32.95	87.53	100.00	0.51	32.95	54.58	12.47									
953E 140	2.7	1.37	5.72	14.15	27.55	33.34	88.33	100.00	0.32	33.34	54.99	11.67									
953E 151	4.1	1.15	4.35	10.41	19.95	24.92	89.04	100.00	1.00	24.92	64.12	10.96									





Hole/Depth	CaCO3 Equiv.	Cumulative - Percentages - - - - -																
		Sand (Sieve intervals) - - - - -																
		10-20	10-40	10-60	10-140	10-230	%Sand + Silt	%Sand, %Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay						
953E 160	4.0	1.16	3.94	10.36	22.58	30.63	88.00	100.00	0.31	30.63	57.37	12.00						
956E 10	6.6	3.06	8.65	15.55	32.57	39.10	86.34	100.00	3.62	39.10	47.24	13.66						
956E 20	5.1	1.68	5.14	9.96	23.14	31.17	88.28	100.00	2.09	31.17	57.11	11.72						
956E 30	4.6	1.69	5.24	9.67	22.74	30.06	87.78	100.00	1.41	30.06	57.72	12.22						
956E 40	5.2	2.07	5.37	11.53	24.86	31.92	93.25	100.00	0.90	31.92	61.33	6.75						
956E 50	4.0	3.45	8.55	14.19	27.73	34.40	92.90	100.00	2.82	34.40	58.49	7.10						
956E 60	4.4	2.65	7.58	13.65	27.89	34.67	97.32	100.00	2.00	34.67	62.64	2.68						
956E 70	4.0	2.75	9.15	18.50	32.43	38.41	92.37	100.00	2.83	38.41	53.96	7.63						
956E 80	3.8	2.16	6.64	13.09	26.99	34.14	93.09	100.00	0.89	34.14	58.95	6.91						
956E 90	4.1	2.21	6.39	14.62	29.44	36.22	91.62	100.00	1.20	36.22	55.39	8.38						
956E 100	4.8	2.06	5.77	12.96	29.09	36.35	72.31	100.00	1.50	36.35	35.96	27.69						
956E 110	5.2	1.75	5.26	11.47	25.82	34.40	90.81	100.00	1.39	34.40	56.41	9.19						
956E 120	4.1	1.47	4.83	11.29	25.00	32.94	76.03	100.00	2.10	32.94	43.08	23.97						
956E 130	4.7	1.48	4.38	9.70	21.92	27.54	83.35	100.00	1.30	27.54	60.81	11.65						
956E 140	4.3	1.61	4.66	10.39	24.22	32.13	93.10	100.00	9.41	32.13	60.98	6.90						
956E 150	4.3	2.00	5.32	11.08	23.89	32.02	90.62	100.00	1.50	32.02	58.60	9.38						
956E 160	4.4	1.58	4.84	13.19	24.86	31.99	90.97	100.00	1.20	31.99	58.98	9.03						
956E 170	3.6	2.32	7.20	13.98	28.01	36.09	90.50	100.00	1.40	36.09	54.42	9.50						
956E 200	2.9	2.40	7.15	16.36	31.68	38.93	90.30	100.00	6.90	38.93	51.37	9.70						
956E 210	3.5	2.52	7.99	15.93	29.84	36.61	90.12	100.00	1.91	36.61	53.51	9.88						
956E 220	2.9	1.94	5.85	14.68	30.18	37.35	89.55	100.00	2.68	37.35	52.20	10.45						
956E 233	3.8	1.90	6.48	14.18	28.87	36.42	95.12	100.00	0.55	36.42	58.70	4.88						
956E 240	4.4	2.34	6.64	14.11	29.42	36.91	88.70	100.00	1.51	36.91	51.78	11.30						



Hole/Depth	CaCO3	Cumulative - Percentages - - - - -										- - - - -					
	Eqiv.	Sand (Sieve intervals) - - - - -										%Sand + Silt	%Sand, %Silt & Clay	%Coarser than 10 mesh	%Sand	%Silt	%Clay
		10-20	20-40	40-60	60-80	80-100	100-140	140-200	200-250	250-300	300-350	350-400	400-450	450-500	500-550	550-600	600-650
956E 250	2.6	2.80	8.75	17.35	33.05	40.32	42.59	45.26	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21
956E 260	2.2	6.62	13.73	23.07	38.52	45.26	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21	81.87	85.53
956E 270	2.2	3.08	9.47	19.24	35.25	42.93	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21	81.87	85.53
956E 280	2.5	2.70	8.65	16.77	34.16	42.59	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21	81.87	85.53
956E 290	2.2	2.48	7.72	15.45	27.90	33.44	42.59	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21	81.87
956E 300	2.1	2.20	7.91	16.52	32.41	39.11	45.26	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21	81.87
957E 20	6.9	2.30	7.26	15.55	29.82	36.39	42.59	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21	81.87
957E 30	6.4	2.28	6.97	12.74	25.75	31.52	38.42	45.38	52.34	59.30	66.26	73.22	80.18	87.14	94.10	101.06	108.02
957E 40	5.2	3.08	8.22	13.10	24.14	26.63	31.52	36.49	41.38	46.27	51.16	56.05	60.94	65.83	70.72	75.61	80.50
957E 50	3.8	2.45	9.04	18.74	36.69	45.38	52.34	59.30	66.26	73.22	80.18	87.14	94.10	101.06	108.02	115.08	122.04
957E 60	0.3	2.44	8.12	17.71	35.09	42.54	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21	81.87	85.53
957E 70	1.0	1.98	6.86	15.88	30.56	38.37	45.38	52.34	59.30	66.26	73.22	80.18	87.14	94.10	101.06	108.02	115.08
957E 80	2.6	1.20	4.74	9.82	25.44	35.42	42.59	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21	81.87
958E 10	4.8	2.91	8.21	16.85	28.85	34.42	42.59	48.93	52.59	56.25	59.91	63.57	67.23	70.89	74.55	78.21	81.87
958E 20	2.9	0.88	3.34	8.11	16.27	21.22	26.17	31.12	36.07	41.02	45.97	50.92	55.87	60.82	65.77	70.72	75.67
958E 30	3.6	1.31	4.53	11.49	19.93	23.44	26.17	28.85	31.52	34.20	36.87	39.55	42.22	44.89	47.57	50.24	52.91
958E 40	4.6	0.41	1.24	1.37	5.81	10.42	15.03	19.64	24.25	28.85	33.46	38.07	42.68	47.29	51.89	56.50	61.11
958E 50	5.1	2.10	6.10	13.57	27.59	30.21	32.83	35.45	38.07	40.69	43.31	45.93	48.55	51.17	53.79	56.41	59.03
958E 57	6.9	0.45	2.17	16.77	45.38	57.05	68.72	80.39	92.06	103.73	115.40	127.07	138.74	150.41	162.08	173.75	185.42



### APPENDIX 3

Grain lithology and bulk chemical analyses for all analyzed samples.

Samples are identified by drill hole number and depth in feet.





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
434E 5	0.0	23.3	4.7	0.0	72.0	0.19	0.81	7.43	33.58	2.05	3.08	0.41	3.57
434E 10	0.0	47.5	1.0	0.5	51.0	0.15	0.87	7.41	33.64	2.11	2.72	0.39	3.89
434E 15	0.0	48.5	5.0	1.0	45.5	0.14	0.87	7.41	33.37	2.16	2.94	0.44	4.02
434E 20	0.0	52.5	0.5	1.0	46.0	0.09	1.01	7.52	33.35	2.12	2.86	0.43	3.76
434E 25	0.0	51.0	4.0	3.0	42.0	0.0	0.78	7.78	33.08	2.14	2.80	0.45	4.34
434E 30	1.7	52.2	5.2	1.7	39.2	0.0	1.03	7.36	33.27	2.21	3.23	0.39	3.91
434E 35	3.5	73.0	3.5	3.0	17.0	0.0	0.64	7.76	33.05	2.13	3.36	0.35	4.15
434E 45	4.0	71.5	3.5	2.0	19.0	0.11	0.98	7.37	33.31	2.14	3.10	0.44	3.91
434E 50	1.0	73.0	1.0	1.0	24.0	0.0	1.20	7.30	33.32	2.15	3.01	0.43	3.55
434E 55	1.0	64.0	2.0	0.0	33.0	0.31	0.69	6.67	34.76	1.93	2.39	0.41	2.72
434E 60	0.0	46.5	16.5	0.0	37.0	0.0	0.51	7.19	35.17	2.01	1.52	0.54	3.54
434E 135	4.2	42.3	4.2	2.1	47.1	0.50	0.66	6.94	34.62	2.09	2.19	0.43	3.32
434E 140	2.6	35.9	5.1	13.8	42.6	0.39	0.85	7.37	33.24	2.18	3.04	0.49	3.72
434E 145	5.8	41.1	4.2	1.6	47.4	0.16	0.75	7.85	33.29	2.01	3.01	0.41	3.76
434E 150	0.5	22.7	5.3	4.3	67.1	0.0	0.84	7.47	33.23	2.00	3.23	0.54	3.89
434E 155	2.5	62.9	6.6	1.5	26.4	0.0	0.85	7.53	33.18	2.08	3.26	0.40	4.10
435E 10	7.8	51.0	2.9	1.5	36.9	0.0	0.61	7.65	33.94	2.29	2.21	0.37	3.85
435E 20	0.0	51.6	2.8	2.0	43.7	0.32	1.07	7.67	33.15	2.32	2.79	0.41	3.63
435E 30	0.0	57.1	3.0	3.6	36.3	0.16	0.71	7.77	33.26	2.21	2.49	0.42	4.15
435E 40	0.0	53.0	4.6	2.7	39.7	0.14	1.08	7.61	32.71	2.15	3.45	0.49	3.95
435E 50	4.0	67.2	8.1	1.5	19.2	0.86	1.42	7.63	32.51	2.14	3.18	0.46	3.42
435E 55	1.0	43.4	4.9	3.9	46.8	0.15	0.99	7.64	32.90	2.31	3.20	0.48	3.83
435E 60	0.4	39.6	2.6	6.5	50.9	0.40	0.95	6.85	33.97	2.22	2.99	0.43	3.24



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)									
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe		
435E 70	1.5	74.9	5.5	1.5	16.6	0.0	0.74	6.98	34.84	2.11	2.41	0.35	3.30		
435E 80	2.0	77.0	3.5	2.0	15.5	0.65	0.74	7.53	33.26	2.39	2.51	0.50	3.84		
435E 90	2.9	66.0	1.4	1.4	28.2	0.15	0.61	7.66	33.85	2.07	2.16	0.46	4.09		
435E 100	7.1	38.4	2.0	8.1	44.4	0.51	0.81	7.61	33.60	2.30	2.04	0.51	3.64		
435E 110	2.1	45.8	1.6	10.9	39.6	0.99	0.78	7.78	33.02	2.35	2.32	0.47	3.75		
435E 120	3.8	34.9	3.2	5.9	52.2	0.65	0.72	7.70	33.69	2.22	1.85	0.43	3.87		
435E 130	5.5	60.8	2.0	0.5	31.2	0.19	0.61	7.57	34.26	2.12	1.72	0.48	3.90		
435E 140	25.9	43.4	1.0	0.0	29.8	0.42	0.65	7.56	34.16	2.02	1.60	0.55	3.92		
435E 160	5.2	65.5	3.6	1.5	24.2	0.53	0.54	7.36	34.19	2.20	1.69	0.52	3.97		
435E 170	6.0	46.3	1.5	1.0	45.3	0.0	0.41	7.37	34.67	1.90	1.82	0.43	4.11		
435E 172	7.0	50.5	3.5	1.0	38.0	0.11	0.51	7.28	34.72	1.90	1.78	0.50	3.89		
435E 180	5.7	31.4	3.6	3.1	56.2	0.0	0.51	7.64	33.96	1.97	1.76	0.68	4.22		
435E 190	6.0	33.7	2.5	11.1	46.7	0.0	0.42	7.36	34.77	1.92	1.62	0.48	4.08		
435E 200	4.5	42.0	2.0	8.5	43.0	0.0	0.58	7.08	34.66	1.75	2.58	0.39	3.72		
435E 210	1.5	37.4	1.5	4.6	54.9	0.0	0.83	8.00	32.77	2.04	2.63	0.46	4.56		
435E 220	3.6	42.7	4.1	4.5	45.0	0.0	0.60	8.47	33.63	1.91	1.70	0.44	3.97		
435E 250	4.3	34.6	4.7	1.9	54.5	0.0	0.60	8.58	32.65	2.09	1.82	0.52	4.76		
435E 252	1.6	37.5	5.2	3.1	52.6	0.0	0.71	8.59	32.72	2.01	2.02	0.55	4.48		
435E 260	2.5	32.5	3.5	4.0	57.5	0.0	0.66	8.41	32.94	2.04	2.25	0.52	4.41		
435E 270	0.0	27.3	0.0	4.5	68.2	0.35	0.75	9.19	31.80	2.03	1.99	0.39	5.08		
435E 290	2.5	29.4	1.5	14.7	52.0	0.0	0.67	7.54	33.94	1.94	2.76	0.54	3.58		
435E 310	2.6	26.1	1.7	43.5	26.1	0.28	0.85	8.42	32.77	1.78	2.60	0.47	4.36		
435E 320	4.8	27.5	4.8	29.5	33.3	0.0	0.66	8.27	32.82	1.90	2.59	0.46	4.64		
435E 330	1.5	48.8	0.0	41.8	8.0	0.37	0.64	8.49	32.60	2.01	2.39	0.48	4.40		
435E 340	4.5	33.8	10.4	5.0	46.3	0.27	0.84	8.31	32.51	1.89	2.55	0.45	4.64		



Hole/Depth Grain Lithology of 1-2mm Fraction (%) Bulk Chemistry of Silt and Clay Fraction (%)

Hole/Depth	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
435E 345	4.6	34.7	4.6	4.6	51.5	0.0	0.63	8.32	32.95	1.86	2.61	0.53	4.32
436E 10	5.3	74.7	4.7	3.2	12.1	0.0	0.68	7.74	33.41	1.92	2.50	0.49	4.46
436E 20	7.2	46.2	3.8	6.7	36.1	0.23	0.95	7.62	33.52	1.79	2.64	0.46	3.90
436E 30	11.1	51.0	4.3	4.3	29.3	0.11	0.90	7.51	33.50	1.78	2.68	0.44	4.05
436E 40	2.5	46.8	3.9	3.0	43.8	0.17	1.11	7.74	32.69	1.82	3.56	0.39	4.02
436E 50	1.0	72.2	1.0	1.0	24.7	0.23	1.07	7.56	33.20	1.91	3.23	0.41	3.61
436E 60	0.5	54.5	1.9	4.8	38.3	0.28	1.20	7.27	33.61	1.92	3.06	0.39	3.37
436E 70	2.0	39.6	3.0	4.6	50.8	0.0	1.02	7.43	33.51	1.98	2.96	0.43	3.86
436E 80	3.3	30.9	0.0	2.8	63.0	0.20	1.15	7.43	33.42	1.90	2.97	0.45	3.70
436E 85	2.1	50.3	2.1	3.7	41.9	0.37	1.10	7.63	32.25	1.77	3.58	0.50	4.43
436E 90	1.5	45.5	0.0	3.5	49.5	0.18	0.76	8.34	32.81	1.72	2.65	0.50	4.31
436E 95	0.5	38.4	3.4	7.4	50.2	0.0	0.68	7.94	33.46	1.84	2.52	0.49	4.15
436E 100	2.5	49.7	5.0	0.5	42.2	0.55	1.11	8.56	31.49	1.69	3.72	0.49	4.23
436E 110	2.0	36.5	6.9	1.5	53.2	0.15	0.87	7.77	33.49	1.82	2.60	0.46	3.94
436E 130	2.5	33.5	19.8	3.0	41.1	0.0	0.97	7.82	33.47	1.76	2.45	0.62	3.87
436E 150	2.5	38.0	3.5	0.5	55.5	0.30	1.08	7.50	33.18	2.01	3.03	0.37	3.90
436E 195	18.9	38.8	5.0	0.0	37.3	0.0	0.86	7.23	33.84	1.92	3.06	0.42	3.80
437E 5	7.0	57.3	3.0	4.0	28.6	0.22	0.89	7.42	33.41	1.97	3.14	0.42	3.83
437E 35	5.5	53.5	3.5	2.5	35.0	0.16	0.87	7.47	33.28	1.96	3.15	0.46	3.99
437E 45	5.7	66.8	2.4	3.3	21.8	0.39	0.98	7.48	33.43	1.86	2.92	0.44	3.73
437E 55	5.6	36.3	1.4	6.5	50.2	0.18	1.05	7.89	32.63	1.88	3.49	0.41	3.97
437E 65	11.5	41.0	4.6	7.8	35.0	0.0	0.99	7.18	34.11	1.90	2.88	0.47	3.44
437E 75	2.5	52.2	6.4	6.9	32.0	0.0	0.53	7.66	34.08	1.82	2.20	0.49	4.10



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
437E 85	2.5	49.5	3.0	0.5	44.5	0.0	0.72	8.25	33.39	1.79	2.31	0.53	4.00
437E 95	6.0	44.8	3.5	0.5	45.3	0.0	0.71	7.67	34.04	1.72	2.22	0.51	3.98
437E 105	11.3	48.0	3.9	2.0	34.8	0.0	0.61	7.45	34.10	1.86	2.10	0.50	4.31
437E 117	4.9	49.5	4.4	0.5	40.7	0.20	0.73	7.96	33.24	1.68	2.70	0.48	4.19
437E 125	11.0	41.0	5.0	2.0	41.0	0.13	0.73	7.79	33.53	1.72	2.36	0.49	4.32
437E 140	1.4	40.4	5.8	0.0	52.4	0.0	0.56	7.58	34.28	1.80	1.87	0.45	4.27
437E 150	3.5	44.6	5.0	0.0	47.0	0.10	0.97	7.17	34.07	1.73	2.78	0.44	3.71
437E 160	2.0	42.9	5.4	0.0	49.8	0.17	0.88	7.55	33.54	1.81	2.74	0.49	3.95
437E 170	1.5	35.1	3.9	27.3	32.2	0.44	1.07	7.72	32.86	2.02	2.99	0.49	3.78
437E 180	4.7	37.3	3.3	12.3	42.5	0.23	0.71	8.22	32.87	1.79	2.78	0.50	4.18
437E 190	3.0	45.5	7.5	1.5	42.5	0.55	0.68	7.88	33.25	1.94	2.21	0.50	4.15
437E 200	5.0	18.5	8.5	42.5	25.5	0.0	0.38	7.42	34.82	1.98	1.43	0.52	4.07
437E 210	4.5	39.8	3.5	13.4	38.8	0.89	1.16	7.39	32.22	2.32	3.86	0.42	3.69
437E 232	17.6	29.0	5.2	17.6	30.6	0.56	1.36	7.45	32.40	2.38	3.72	0.39	3.55
437E 240	3.5	39.2	1.5	7.5	48.2	0.0	0.44	7.87	33.56	2.01	2.20	0.52	4.31
437E 260	0.5	15.9	0.5	47.5	35.4	0.74	0.70	7.93	32.76	2.03	2.58	0.53	4.07
439E 5	1.4	40.9	0.9	9.8	47.0	0.59	1.12	6.76	33.56	1.91	3.82	0.42	3.13
439E 10	2.0	33.5	2.0	5.9	56.7	0.34	0.78	7.85	33.00	1.96	2.72	0.47	4.06
439E 20	4.5	38.5	5.5	1.0	50.5	0.45	0.83	7.58	33.19	2.06	2.82	0.42	4.03
439E 30	8.8	31.2	3.9	4.4	51.7	0.58	0.86	8.32	31.89	2.01	3.13	0.52	4.47
439E 40	3.4	40.4	1.0	3.0	52.2	0.51	0.96	7.83	33.04	2.25	2.79	0.41	3.50
439E 50	5.0	33.7	5.0	6.0	50.3	0.67	1.11	7.62	32.18	2.03	3.59	0.43	4.22
439E 60	1.5	42.1	3.0	4.0	49.5	0.78	1.13	7.49	32.13	2.45	3.49	0.47	4.04
439E 70	2.2	38.5	3.8	4.9	50.5	0.84	0.95	7.56	32.98	2.53	2.74	0.34	3.64





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)				Bulk Chemistry of Silt and Clay Fraction (%)									
	Local	Acid	Basic	Carb. Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe		
439E 80	0.9	41.7	1.4	4.7	51.2	1.40	1.03	7.35	31.94	2.63	3.06	0.49	4.01	
439E 90	3.3	43.6	4.7	4.7	43.6	1.74	0.87	7.60	31.93	2.81	2.81	0.36	3.78	
439E 100	5.0	59.9	1.5	2.0	31.7	0.78	1.02	7.31	31.75	2.54	4.79	0.46	3.56	
439E 108	3.0	39.8	1.5	31.8	23.9	0.91	0.97	7.50	32.73	2.76	2.66	0.43	3.78	
439E 118	3.4	51.2	2.5	5.9	36.9	0.71	0.55	8.38	33.47	2.34	1.30	0.35	3.96	
440E 10	1.5	62.6	2.0	1.5	32.3	0.32	0.74	6.01	35.50	1.90	2.79	0.39	3.06	
440E 20	7.5	39.8	4.5	2.0	46.3	1.05	0.76	7.50	33.33	2.47	2.15	0.37	3.73	
440E 30	5.1	41.1	2.0	6.1	45.7	1.18	0.83	7.56	32.96	2.38	2.31	0.38	3.89	
440E 40	4.4	37.6	2.0	7.3	48.8	1.49	0.75	7.73	32.20	2.67	2.41	0.43	4.15	
440E 50	3.8	33.2	4.8	8.7	49.5	0.37	1.02	7.11	33.40	2.30	3.20	0.35	3.69	
440E 60	1.4	50.2	0.9	5.5	42.0	0.63	0.62	7.83	33.91	2.13	1.90	0.46	3.47	
440E 70	3.4	44.7	1.5	3.9	46.6	1.09	0.73	8.08	33.35	2.10	1.59	0.38	3.76	
440E 80	7.8	36.6	4.9	2.0	48.8	0.78	0.74	7.73	33.32	2.14	2.10	0.43	3.94	
440E 90	3.3	42.5	2.8	0.5	50.9	1.88	0.71	8.07	32.00	2.49	2.03	0.47	4.11	
440E 100	7.8	33.0	3.9	1.0	54.4	0.70	0.54	8.80	33.21	2.22	1.59	0.41	3.55	
440E 110	7.5	31.8	0.0	3.7	57.0	0.64	0.72	7.95	33.17	2.30	2.19	0.46	3.80	
440E 120	11.6	29.5	0.5	3.7	54.7	0.97	0.65	7.98	33.13	2.18	1.49	0.51	4.41	
440E 150	7.1	45.3	2.8	0.5	44.3	0.86	0.41	7.72	33.73	2.30	1.78	0.45	3.92	
440E 170	19.2	21.2	3.0	0.0	56.6	0.98	0.42	7.40	34.07	2.24	1.50	0.46	4.03	
440E 180	15.4	30.0	2.4	1.2	51.0	1.49	0.54	7.64	33.13	2.61	1.43	0.51	4.06	
440E 190	18.4	35.9	1.9	1.9	41.7	1.49	0.63	7.91	32.71	2.47	1.65	0.51	4.12	
440E 210	12.6	32.3	0.0	0.4	54.7	1.15	0.65	7.86	32.66	2.25	2.27	0.43	4.20	
440E 219	13.1	28.3	0.0	1.0	57.6	0.0	0.47	7.35	35.10	1.82	1.46	0.47	3.79	
440E 230	10.1	28.5	1.8	5.7	53.9	1.77	0.69	8.15	31.57	2.69	2.17	0.47	4.51	



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)				Bulk Chemistry of Silt and Clay Fraction (%)									
	Local	Acid	Basic	Carb. Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe		
440E 240	4.7	44.3	2.6	5.2	43.2	0.12	0.69	7.75	33.77	1.85	2.29	0.46	4.07	
443E 250	8.3	42.2	2.9	5.3	41.3	0.87	0.43	8.32	33.38	2.14	1.65	0.48	3.85	
440E 256	7.6	32.2	1.4	5.7	53.1	1.50	0.77	8.31	31.20	2.74	2.23	0.48	4.69	
455E 5	4.2	34.9	0.0	19.1	41.9	0.60	0.57	7.32	34.67	2.03	1.56	0.45	3.53	
455E 10	4.3	73.5	3.7	3.7	14.8	1.23	0.77	7.77	32.50	2.71	1.71	0.40	4.50	
455E 15	6.2	66.4	2.7	0.9	23.9	1.09	0.88	7.83	33.20	2.33	2.14	0.41	3.34	
455E 20	7.3	41.7	1.0	5.7	44.3	1.31	0.94	8.09	32.21	2.59	2.16	0.50	3.82	
455E 35	1.9	31.5	2.8	20.4	43.5	1.12	1.85	7.50	31.10	2.62	4.23	0.32	3.58	
455E 50	2.4	62.3	9.0	1.4	25.0	1.21	1.34	7.55	32.06	2.64	3.18	0.34	3.68	
455E 65	2.0	75.0	6.5	1.0	15.5	2.01	1.95	8.04	29.13	3.01	4.60	0.37	4.03	
455E 80	20.3	55.0	7.9	2.0	14.9	0.24	1.93	7.17	31.48	2.19	5.16	0.35	3.72	
455E 95	7.3	49.7	2.6	1.6	38.9	2.29	1.44	7.80	30.10	3.18	3.62	0.36	4.01	
455E 110	6.6	54.3	5.6	2.5	31.0	0.49	1.58	8.31	31.16	2.14	3.58	0.41	4.31	
456E 5	1.4	44.6	1.4	16.2	36.5	0.0	0.68	7.66	33.96	1.99	2.01	0.45	4.19	
456E 15	23.5	41.2	0.0	0.0	35.3	0.26	0.87	7.75	33.02	2.12	2.39	0.54	4.45	
456E 25	0.0	38.7	0.0	11.7	49.5	0.0	0.66	7.43	34.08	1.92	2.12	0.45	4.16	
456E 35	5.3	43.8	0.0	10.1	40.8	0.0	0.72	7.55	33.79	2.00	2.41	0.38	4.24	
456E 40	12.5	59.0	4.0	2.5	22.0	0.0	0.58	8.01	33.97	1.83	2.01	0.47	3.93	
456E 45	6.9	49.8	6.9	5.4	31.0	0.0	0.52	6.88	35.24	1.88	2.26	0.32	3.48	
456E 55	22.2	47.7	6.8	1.7	21.6	0.0	1.00	8.25	32.56	2.18	2.63	0.42	4.43	
456E 65	21.0	56.5	4.5	1.5	16.5	0.0	1.07	8.77	32.24	2.07	2.41	0.41	4.44	
456E 75	12.2	55.8	3.4	3.4	25.2	0.62	1.35	7.81	32.24	2.20	3.27	0.39	3.87	
456E 85	5.0	80.5	2.5	2.0	10.0	0.25	1.03	7.90	32.91	2.12	2.77	0.43	3.98	



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Bg	Al	Si	K	Ca	Ti	Fe
456E 95	0.7	54.9	4.6	2.6	37.3	0.0	1.09	8.35	32.69	2.10	2.48	0.49	4.13
456E 120	0.0	72.2	2.6	2.1	23.2	0.54	2.29	7.25	30.47	2.19	5.58	0.44	3.91
456E 125	0.0	79.5	4.0	1.0	15.5	0.71	2.31	7.02	30.57	2.14	5.68	0.47	3.81
456E 130	0.5	35.7	1.0	4.8	58.0	0.22	0.66	7.36	34.04	2.01	2.35	0.47	3.90
456E 135	2.0	62.0	5.5	1.5	29.0	0.51	0.63	6.70	34.67	1.87	2.20	0.49	3.65
456E 145	0.8	40.6	3.1	0.8	54.7	0.0	0.28	7.62	34.85	1.89	1.37	0.43	4.12
456E 150	4.0	61.5	2.5	1.0	31.0	0.25	0.42	7.67	34.32	2.09	1.42	0.42	4.23
457E 10	1.0	80.5	1.5	1.0	16.0	0.77	1.14	6.52	33.94	1.99	3.23	0.36	3.36
457E 30	12.7	58.8	4.9	1.0	22.5	0.61	1.28	6.87	33.15	2.07	3.56	0.40	3.63
457E 50	0.0	65.1	5.3	0.6	29.0	0.50	1.22	7.32	32.84	2.09	3.43	0.36	3.85
457E 60	0.5	75.5	9.0	3.5	11.5	0.48	2.36	6.64	30.84	2.07	6.03	0.37	3.81
457E 70	0.0	67.3	1.5	2.0	29.2	0.72	2.38	6.43	31.15	2.04	6.07	0.37	3.43
457E 87	5.1	50.0	10.1	14.1	20.7	0.52	1.88	6.54	31.88	2.08	5.47	0.43	3.45
457E 100	0.5	62.6	4.4	1.9	30.6	0.0	1.12	7.48	32.82	2.37	3.32	0.41	4.06
457E 110	6.0	74.5	1.0	4.0	14.5	0.0	0.51	6.20	35.53	1.95	2.75	0.42	3.30
457E 120	5.6	56.1	2.3	3.7	32.2	0.26	0.59	8.86	33.14	1.77	2.18	0.54	3.53
457E 130	5.0	69.0	4.5	3.0	18.5	0.13	1.18	7.27	32.99	2.21	3.28	0.40	4.10
457E 185	1.0	48.8	2.5	1.5	46.3	0.43	1.03	7.32	33.01	2.07	3.14	0.46	4.07
457E 200	0.0	64.5	4.4	3.4	27.6	0.18	1.14	7.41	33.13	2.06	3.09	0.36	4.05
457E 210	12.5	61.5	8.5	1.5	16.0	0.64	2.06	6.78	32.06	1.97	4.96	0.38	3.18
457E 240	1.0	62.0	1.0	1.0	35.0	0.44	1.57	7.00	32.32	2.10	4.38	0.40	3.71
458E 15	0.0	56.5	1.0	24.5	18.0	0.43	1.51	7.61	31.52	2.23	4.27	0.39	4.21
458E 25	1.5	66.5	1.0	9.5	21.5	0.39	0.85	7.72	33.24	1.93	2.58	0.57	3.97





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)									
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe		
458E 80	6.0	66.7	2.5	1.5	23.4	0.14	0.97	7.43	33.07	2.10	3.31	0.47	3.98		
458E 185	1.0	49.8	3.5	8.5	37.3	0.27	0.61	7.16	34.11	2.02	2.47	0.40	4.03		
458E 225	0.5	65.5	3.9	6.4	23.6	0.67	0.96	7.01	33.94	1.92	2.74	0.42	3.48		
458E 235	1.0	51.0	5.2	3.1	39.6	0.28	0.86	5.95	35.15	1.82	2.75	0.46	3.57		
458E 255	3.0	68.0	1.5	7.4	20.2	0.47	0.85	8.28	32.14	4.93	2.56	0.25	2.64		
459E 10	1.5	58.7	3.0	6.5	30.3	0.21	1.44	7.65	32.66	2.09	3.08	0.46	3.91		
459E 40	1.0	73.6	1.0	0.0	24.4	0.36	1.36	7.53	32.44	2.17	3.41	0.33	3.97		
459E 60	0.7	71.9	7.9	0.0	19.4	0.38	1.28	6.80	33.76	2.00	3.10	0.41	3.53		
459E 70	1.0	68.0	2.5	8.9	19.7	0.31	2.42	6.79	31.10	2.07	5.96	0.38	3.44		
459E 100	1.0	70.4	3.0	4.5	21.1	0.74	1.55	7.94	30.93	2.17	3.82	0.47	4.72		
461E 5	1.0	56.3	2.0	2.5	38.1	0.76	0.83	6.77	33.75	1.93	2.49	0.43	4.40		
461E 10	5.5	70.0	2.0	1.5	21.0	0.42	0.80	8.01	33.16	2.02	2.15	0.45	4.23		
461E 15	7.5	71.5	1.5	2.5	17.0	0.12	0.84	7.54	34.00	1.98	2.18	0.43	3.85		
461E 20	0.0	59.0	6.5	4.0	30.5	0.17	0.80	7.43	34.06	1.97	2.19	0.45	3.88		
461E 25	21.0	57.5	10.0	0.5	11.0	0.45	0.92	7.57	33.52	2.05	2.34	0.39	3.96		
461E 30	2.4	52.4	1.6	17.5	26.2	0.0	0.77	7.50	33.99	1.97	2.21	0.40	3.94		
461E 35	1.8	71.6	3.0	3.0	20.7	0.30	0.93	8.12	32.56	2.06	2.56	0.56	4.39		
461E 40	1.2	56.5	4.3	12.4	25.5	0.0	0.78	8.13	32.77	2.18	2.57	0.48	4.51		
461E 45	19.5	53.5	4.0	1.0	22.0	0.40	0.80	7.88	33.25	1.97	2.44	0.45	3.86		
461E 50	7.0	61.1	7.0	2.7	22.2	0.0	0.69	7.37	34.17	1.96	2.19	0.45	4.09		
461E 65	5.5	67.0	3.0	3.0	21.5	0.34	0.88	7.59	33.35	2.01	2.46	0.54	4.09		
461E 69	1.0	54.7	4.4	4.4	35.5	0.30	0.80	7.01	34.11	2.01	2.48	0.44	3.95		
461E 75	2.0	67.0	5.5	1.0	24.5	0.32	0.75	7.22	34.59	1.89	2.00	0.41	3.56		



Hole/Depth Grain Lithology of 1-2mm Fraction (%) Bulk Chemistry of Silt and Clay Fraction (%)

		Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
462E	5	0.0	47.9	1.4	8.6	42.1	0.35	0.65	7.29	35.10	1.82	1.44	0.39	3.43
462E	20	1.6	50.8	3.1	7.3	37.2	0.0	0.87	7.59	33.76	2.09	2.28	0.43	4.05
462E	35	2.4	29.3	6.1	25.6	36.6	0.12	0.84	7.70	33.64	1.90	2.33	0.43	4.11
462E	50	0.0	55.4	5.9	14.9	23.8	0.43	2.41	7.04	30.76	2.13	5.70	0.41	3.70
462E	60	5.5	75.5	3.0	4.0	12.0	0.37	1.47	8.12	31.48	2.17	3.89	0.36	4.16
462E	75	1.0	61.9	2.5	8.4	26.2	0.0	1.97	7.60	31.14	2.26	4.67	0.39	4.13
462E	90	10.5	58.5	4.5	6.5	20.0	0.23	1.31	8.41	31.41	2.10	3.58	0.46	4.33
463E	10	1.0	45.5	11.5	12.0	30.0	0.0	0.71	7.57	33.84	1.95	2.16	0.49	4.29
463E	20	5.5	71.5	3.5	4.5	15.0	0.0	0.46	6.82	36.10	1.66	1.42	0.35	3.31
463E	25	14.9	47.5	4.0	5.0	28.7	0.90	0.59	7.78	33.00	2.25	2.06	0.49	4.33
463E	30	11.0	69.1	3.7	1.5	14.7	0.33	0.74	7.69	33.86	1.96	2.05	0.34	4.02
463E	40	29.5	39.0	3.5	4.0	24.0	0.12	0.98	8.59	32.18	2.05	2.55	0.40	4.65
463E	50	10.0	65.5	4.0	0.5	20.0	0.11	0.84	7.73	33.48	2.02	2.29	0.50	4.18
463E	60	0.5	56.3	2.2	7.7	33.3	0.34	0.91	7.50	33.61	1.92	2.42	0.50	3.93
463E	70	16.0	59.0	3.5	3.5	18.0	0.0	0.90	7.68	33.67	2.06	2.31	0.45	4.01
463E	75	20.0	55.5	3.5	1.0	12.0	0.52	0.95	8.56	32.00	2.24	2.36	0.43	4.61
463E	80	0.0	44.8	2.0	32.3	20.9	1.24	0.75	7.32	33.21	1.86	2.44	0.43	4.23
463E	85	5.0	74.0	3.0	4.0	14.0	0.21	0.79	7.09	34.15	1.98	2.26	0.50	4.04
463E	90	4.0	71.5	3.5	2.5	18.5	0.40	0.91	7.88	33.16	2.04	2.26	0.50	4.02
463E	95	2.0	63.0	9.0	2.0	24.0	0.24	0.90	7.62	33.30	2.06	2.55	0.42	4.22
463E	100	0.0	50.0	5.2	19.0	25.8	0.43	1.08	8.07	32.49	2.17	2.46	0.48	4.35
496E	210	0.0	41.9	5.6	17.9	34.6	0.41	0.89	7.90	32.53	1.98	2.69	0.56	4.48



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)									
	Local Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe			
496E 250	2.5	45.7	1.5	5.5	44.7	0.0	0.58	7.38	34.67	1.93	1.35	0.51	4.08		
496E 290	11.9	51.5	1.0	3.5	32.2	0.12	0.18	6.77	35.89	1.96	1.05	0.49	3.87		
496E 310	3.1	53.1	4.1	3.6	36.2	0.0	0.13	8.21	35.09	1.76	1.00	0.35	3.71		
496E 330	1.6	53.3	3.3	5.7	36.1	0.11	0.27	7.08	35.53	1.87	1.14	0.52	3.85		
496E 345	4.7	66.8	3.2	1.6	23.7	0.0	0.45	7.39	34.82	1.71	1.39	0.53	4.10		
496E 365	21.3	30.2	11.4	3.0	34.2	0.35	0.63	7.66	33.92	1.75	1.85	0.51	4.28		
497E 30	1.0	61.8	4.0	8.5	24.6	0.46	1.61	7.86	31.18	2.18	4.23	0.44	4.24		
497E 40	5.5	68.5	3.5	5.5	17.0	0.28	1.57	7.73	31.74	2.23	3.98	0.47	3.97		
497E 50	18.5	45.5	8.0	5.5	22.5	0.23	1.20	7.44	33.18	2.10	3.15	0.35	3.72		
497E 80	11.9	54.7	5.0	3.0	25.4	0.21	1.03	8.13	32.80	1.95	2.49	0.49	4.23		
497E 90	25.5	48.0	1.0	1.5	24.0	0.25	0.53	8.39	32.78	1.91	2.16	0.48	4.85		
497E 100	30.8	40.3	4.0	4.0	20.9	0.58	0.76	7.78	33.42	1.87	2.54	0.40	3.83		
497E 110	11.5	59.5	3.0	1.0	25.0	0.12	0.84	8.05	33.08	1.99	2.54	0.45	4.18		
497E 120	60.5	26.0	1.0	2.5	10.0	0.17	1.21	7.91	32.54	2.07	3.03	0.41	4.22		
497E 130	0.0	69.0	6.5	1.5	23.0	0.34	1.86	7.91	30.16	2.26	4.92	0.44	4.83		
497E 140	3.0	60.7	3.0	12.4	20.9	0.21	1.87	7.84	31.17	2.23	4.00	0.48	4.21		
497E 150	3.0	67.5	6.0	2.5	21.0	0.56	1.86	7.46	31.21	2.10	4.64	0.34	4.15		
497E 160	6.5	55.7	5.0	8.0	24.9	0.0	1.63	7.73	31.22	2.23	4.56	0.44	4.29		
497E 170	2.3	69.4	1.2	1.7	25.4	0.40	1.93	7.59	31.45	2.12	4.44	0.38	3.69		
497E 220	1.0	65.0	5.0	1.5	27.5	0.21	1.06	8.74	31.65	2.19	2.77	0.47	4.68		
497E 230	4.5	63.5	6.5	5.0	20.5	0.29	1.23	8.92	31.40	2.20	2.89	0.48	4.38		
497E 240	0.7	61.5	2.2	7.4	28.1	0.17	1.20	10.02	29.81	1.49	3.83	0.46	5.09		
497E 250	6.5	64.5	6.5	2.0	20.5	0.12	0.92	7.38	33.68	1.96	3.00	0.39	3.71		
497E 260	0.0	67.0	3.5	3.5	26.0	0.49	1.21	7.57	32.69	2.10	3.04	0.48	4.01		



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)				Bulk Chemistry of Silt and Clay Fraction (%)								
	Local	Acid	Basic	Carb.	Qtz.	Na	Ng	Al	Si	K	Ca	Ti	Fe
497E 270	2.5	58.0	11.0	3.5	25.0	0.67	1.20	8.07	32.05	2.24	3.15	0.43	3.97
497E 280	2.3	63.6	6.1	6.8	21.2	0.26	1.09	9.14	31.19	2.23	2.61	0.50	4.83
497E 290	1.5	66.0	4.0	2.5	26.0	0.37	1.11	8.35	32.21	2.10	2.67	0.50	4.25
497E 295	2.1	70.8	0.0	4.2	22.9	0.23	1.15	8.03	32.56	2.03	2.89	0.47	4.11
498E 15	3.0	60.0	3.0	5.5	28.5	0.0	0.91	8.11	33.49	1.96	2.30	0.53	3.68
498E 25	2.0	54.5	8.1	6.1	29.3	0.0	0.75	7.93	33.35	1.97	2.53	0.45	4.17
498E 40	1.0	61.0	7.0	4.5	26.5	0.0	0.90	7.65	33.65	2.02	2.46	0.44	3.95
498E 55	9.5	63.5	7.0	3.5	16.5	0.0	0.91	7.69	33.97	1.88	2.29	0.43	3.70
498E 65	11.5	52.5	6.0	4.0	26.0	0.0	0.81	7.69	33.76	1.91	2.20	0.45	4.18
498E 70	1.0	69.5	7.0	1.5	21.0	0.0	0.85	7.78	33.44	1.91	2.55	0.44	4.16
498E 100	0.5	67.5	3.5	8.0	20.5	0.34	1.89	8.32	30.66	2.17	4.18	0.50	4.02
498E 110	3.0	69.5	3.5	2.5	21.5	0.27	1.78	8.21	31.14	2.32	3.49	0.42	4.47
498E 120	5.5	50.5	4.5	14.5	25.0	0.33	1.90	7.82	30.88	2.28	4.40	0.43	4.09
498E 130	5.1	70.1	5.1	2.8	16.9	0.20	1.48	7.88	31.55	2.27	3.83	0.50	4.33
498E 140	3.5	69.0	3.5	4.5	19.5	0.18	1.83	7.90	30.92	2.25	4.59	0.37	4.29
498E 150	0.5	63.0	4.0	8.5	24.0	0.32	1.76	7.63	31.83	2.22	3.96	0.44	3.77
498E 160	17.5	48.0	5.5	3.5	25.5	0.15	0.94	8.01	32.82	1.96	2.85	0.47	4.18
498E 170	6.0	59.5	7.0	7.5	20.0	0.0	1.61	7.85	31.59	2.22	4.01	0.33	4.23
498E 180	1.5	69.0	4.5	5.5	19.5	0.0	1.66	7.98	31.40	2.20	4.15	0.43	4.24
498E 190	1.5	53.0	4.5	19.5	21.5	0.13	1.78	7.77	31.39	2.22	4.16	0.50	3.97
498E 200	1.0	69.5	1.0	4.0	24.5	0.0	2.18	7.48	31.17	2.28	4.89	0.38	3.76
498E 250	0.5	59.0	3.5	9.0	28.0	0.70	1.94	7.33	31.43	2.14	4.56	0.39	3.73
498E 265	1.5	66.0	6.0	0.5	26.0	0.39	1.82	7.91	30.95	2.17	4.19	0.44	4.20
498E 270	4.9	56.4	7.4	4.3	27.0	0.68	1.13	7.53	32.70	2.13	2.94	0.53	3.88





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Hg	Al	Si	K	Ca	Ti	
498E 275	0.0	59.0	5.0	3.0	33.0	0.48	1.11	7.22	33.27	1.98	3.06	0.46	
498E 280	4.4	62.2	4.4	2.2	26.7	0.65	0.97	7.27	33.62	1.98	2.80	0.40	
498E 290	0.0	63.5	4.0	5.0	27.5	0.25	0.92	7.48	33.35	2.05	2.87	0.51	
706E 50	6.0	68.5	4.5	3.5	17.5	0.38	1.12	8.21	32.68	1.98	2.98	0.40	
706E 60	3.5	73.0	3.5	1.5	18.5	0.57	1.16	7.91	32.43	1.97	2.81	0.44	
706E 70	4.0	67.5	1.5	2.0	25.0	0.56	1.10	7.75	32.73	1.99	2.91	0.46	
706E 80	5.0	62.5	1.5	5.5	25.5	0.17	0.83	7.95	34.18	1.89	1.61	0.50	
706E 100	3.0	57.0	5.5	2.5	32.0	0.54	1.12	7.03	33.96	1.85	2.87	0.48	
705E 110	1.0	72.5	2.5	4.0	20.0	0.39	0.77	7.91	33.00	4.01	2.54	0.34	
706E 120	6.0	62.5	4.5	4.0	23.0	0.45	1.06	7.77	33.20	1.89	2.84	0.42	
706E 125	2.5	62.0	2.0	4.0	29.5	0.0	0.97	7.29	33.86	1.84	2.86	0.35	
705E 130	0.5	62.0	0.5	7.5	29.5	0.41	1.06	8.42	32.67	2.06	2.82	0.33	
706E 200	0.5	54.0	7.5	6.0	32.0	0.39	1.09	8.06	32.01	2.03	3.35	0.50	
706E 210	2.5	63.5	4.5	2.5	27.0	0.62	1.27	8.81	31.34	2.11	2.67	0.50	
705E 220	4.5	58.5	7.0	4.5	25.5	0.29	1.02	8.17	32.54	1.97	2.50	0.54	
706E 230	7.1	51.5	2.0	5.1	34.3	0.27	0.61	7.50	34.85	1.73	1.34	0.39	
705E 240	4.0	60.0	2.7	1.3	32.0	0.14	0.74	8.39	33.53	2.01	1.77	0.48	
706E 250	9.7	64.6	4.9	2.8	18.1	0.36	0.73	7.53	34.46	1.85	1.62	0.40	
705E 260	53.0	29.0	0.5	3.0	14.5	0.76	0.70	8.01	33.91	1.75	1.93	0.41	
705E 270	8.5	61.0	5.0	3.5	22.0	0.22	0.68	7.34	33.98	1.67	2.58	0.44	
706E 280	14.5	53.5	3.5	2.5	26.0	0.16	0.45	7.51	34.37	1.81	1.78	0.52	
705E 290	6.0	73.0	1.5	1.0	18.5	0.33	0.83	8.26	32.92	1.94	2.01	0.54	
705E 325	8.0	51.0	1.5	9.0	30.5	0.43	0.76	7.91	33.93	1.84	1.73	0.74	
706E 345	45.0	33.6	13.6	1.4	6.4	0.29	0.67	7.16	34.78	1.83	1.67	0.49	



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
706E 348	10.5	62.0	4.5	1.5	21.5	0.71	0.71	7.04	34.68	1.94	1.62	0.46	3.59
706E 360	6.5	70.5	4.0	2.5	16.5	0.44	0.98	8.22	32.65	1.93	2.17	0.53	4.29
709E 10	1.9	66.9	3.8	4.5	22.9	0.25	1.16	7.77	33.00	2.01	2.81	0.53	3.72
709E 20	3.0	56.5	2.0	8.5	30.0	0.39	1.05	8.17	33.25	2.11	2.32	0.45	3.35
709E 30	1.0	72.0	3.0	2.5	21.5	0.38	0.98	7.30	33.87	1.93	2.53	0.50	3.58
709E 40	3.0	67.5	2.0	6.0	21.5	0.29	1.49	8.19	31.28	2.16	3.94	0.49	4.09
709E 50	1.6	63.7	6.3	5.3	23.2	0.28	1.23	7.55	32.66	1.99	3.31	0.46	3.82
709E 60	0.5	68.0	4.0	4.5	23.0	0.20	1.04	7.51	33.37	2.00	3.04	0.39	3.56
709E 70	2.9	65.7	1.5	3.4	26.5	0.34	1.23	7.55	32.54	2.12	3.42	0.38	3.99
709E 80	0.5	64.0	2.0	8.0	25.5	0.39	1.29	7.63	33.14	2.13	2.92	0.35	3.48
709E 90	0.5	71.0	6.0	4.5	18.0	0.55	1.78	7.74	31.68	2.09	3.65	0.48	3.97
709E 100	1.5	62.9	1.5	3.5	30.7	0.25	0.73	7.23	34.57	1.99	1.76	0.45	3.79
709E 110	1.0	67.5	3.0	3.0	25.5	0.0	0.69	7.74	34.73	1.98	1.62	0.45	3.29
709E 120	1.5	60.0	0.0	3.0	35.5	0.40	0.62	7.73	34.46	2.01	1.72	0.44	3.31
709E 130	2.5	66.5	2.5	2.0	26.5	0.08	0.55	7.10	35.37	1.84	1.63	0.46	3.27
709E 140	2.0	66.0	1.0	4.5	26.5	0.26	0.60	6.49	35.83	1.85	1.85	0.40	3.08
709E 150	2.0	61.0	3.5	3.5	30.0	0.50	0.90	7.65	33.41	2.02	2.02	0.49	3.83
709E 160	5.5	58.0	2.5	5.5	20.5	0.0	0.53	7.93	33.71	1.94	2.13	0.50	4.26
709E 170	4.0	56.5	2.0	2.5	35.0	0.72	0.37	6.93	34.89	1.97	1.68	0.41	3.52
710E 5	1.5	67.5	2.0	3.5	25.5	0.40	1.40	8.06	31.08	2.16	3.93	0.51	4.55
710E 90	6.5	65.6	4.3	3.2	20.4	0.38	0.89	6.80	34.98	1.87	2.22	0.39	3.15
710E 95	4.5	69.5	2.0	4.5	19.5	0.24	1.01	7.40	34.04	1.90	2.50	0.38	3.47
710E 110	2.5	63.0	5.5	5.0	24.0	0.54	1.00	7.70	32.77	1.94	3.26	0.41	3.74



Hole/Depth	Grain Lithology of 1-2 $\mu$ Fraction (%)				Bulk Chemistry of Silt and Clay Fraction (%)									
	Local	Acid	Basic	Carb. Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe		
710E 115	1.5	62.5	7.0	2.5	26.5	0.48	1.25	7.83	31.73	2.17	3.65	0.52	4.11	
710E 120	1.5	67.5	5.5	2.5	23.0	0.34	0.85	7.63	34.24	1.85	2.20	0.38	3.29	
710E 125	2.5	69.0	4.5	2.5	21.5	0.0	0.78	7.70	34.08	1.82	2.58	0.42	3.26	
710E 130	2.0	65.5	5.5	4.0	23.0	0.19	0.94	7.86	33.20	1.94	2.93	0.44	3.54	
710E 135	2.0	66.7	6.1	6.1	19.2	0.25	0.87	7.55	33.92	1.88	2.50	0.45	3.43	
710E 140	2.0	73.5	5.0	2.5	17.0	0.64	1.41	7.43	32.77	2.11	3.19	0.46	3.54	
710E 145	0.5	78.0	2.5	2.5	16.5	0.46	0.94	7.58	33.60	2.28	2.35	0.39	3.57	
710E 150	2.5	69.5	4.0	2.5	21.5	0.56	1.10	7.53	32.75	1.96	3.11	0.43	3.85	
710E 155	1.5	67.0	5.5	1.5	24.5	0.33	1.19	7.65	32.88	2.12	2.84	0.43	3.86	
710E 160	0.0	65.5	5.0	4.5	25.0	0.55	1.11	7.56	33.37	1.94	2.62	0.39	3.53	
710E 165	2.0	66.5	2.5	3.5	25.5	0.30	0.83	7.67	34.27	2.01	2.11	0.34	3.24	
710E 244	2.0	63.0	1.5	5.0	28.5	0.57	1.23	7.51	32.62	2.00	3.18	0.51	3.76	
710E 273	9.5	84.1	1.0	3.0	2.5	0.26	1.03	7.89	33.10	1.94	2.72	0.41	3.83	
713E 60	16.2	62.2	6.3	2.7	12.6	0.80	0.89	7.86	32.80	2.10	2.64	0.39	4.03	
713E 70	12.0	66.0	8.5	4.5	9.0	0.31	0.94	8.62	33.04	1.92	1.50	0.40	4.26	
713E 80	1.5	77.5	3.0	2.0	16.0	0.61	1.39	7.65	32.21	2.11	3.11	0.46	4.10	
713E 90	13.0	67.5	2.5	3.0	14.0	0.54	1.15	7.29	34.05	1.98	2.54	0.31	3.10	
713E 100	1.0	72.5	6.0	1.5	19.0	0.33	0.65	7.69	34.43	1.88	2.11	0.38	3.15	
713E 110	1.0	71.0	2.5	4.5	21.0	0.31	0.83	8.04	34.02	1.85	1.99	0.35	3.37	
713E 120	0.0	76.0	5.0	1.5	17.5	0.41	1.38	7.01	33.51	1.96	3.25	0.37	3.41	
714E 5	5.5	68.5	2.0	6.5	17.5	0.0	1.87	7.55	31.68	2.05	4.64	0.42	3.76	
714E 10	0.5	77.0	3.5	2.0	17.0	0.49	1.71	7.83	31.23	2.30	4.52	0.35	3.78	
714E 15	0.0	71.5	5.0	4.0	19.5	0.36	2.03	7.52	31.37	2.07	4.74	0.38	3.68	



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)								
	Local	Acid	Basic	Carb.	Otz.	Na	Mg	Al	Si	K	Ca	Ti	Fe	
714E 20	3.0	72.5	2.5	3.5	18.5	0.0	1.80	7.87	31.62	2.10	4.44	0.35	3.74	
714E 25	4.5	65.5	3.0	8.5	18.5	0.0	1.87	7.36	31.38	2.20	5.25	0.39	3.79	
714E 30	7.0	76.0	3.5	3.0	10.5	0.28	1.89	8.02	30.90	2.23	4.35	0.48	4.02	
714E 35	1.5	66.5	13.0	3.5	15.5	0.42	1.81	7.75	31.41	2.20	4.40	0.42	3.68	
714E 40	0.5	66.5	8.5	2.5	22.0	0.23	1.77	7.69	31.46	2.22	4.49	0.37	3.75	
714E 45	3.0	57.5	5.0	7.0	27.5	0.42	1.76	7.88	31.29	2.28	4.47	0.39	3.66	
714E 50	6.5	67.0	4.5	3.5	18.5	0.46	2.02	7.62	31.10	2.16	4.67	0.43	3.78	
714E 55	3.5	74.0	4.5	5.0	13.0	0.23	1.61	7.99	31.39	2.18	4.40	0.42	3.84	
714E 60	0.5	70.5	7.0	3.0	19.0	0.59	2.22	5.91	32.59	1.86	5.42	0.33	2.98	
714E 65	2.2	69.8	3.3	3.8	20.9	0.32	1.72	7.96	31.36	2.10	4.17	0.39	3.93	
714E 70	3.5	67.5	3.0	8.5	17.5	0.53	1.79	7.61	31.52	2.17	4.33	0.44	3.71	
714E 75	3.0	65.5	5.0	2.5	24.0	0.0	1.82	7.81	31.54	2.09	4.40	0.48	3.82	
714E 80	1.5	75.5	3.0	4.0	16.0	0.30	1.77	8.04	31.18	2.08	4.45	0.40	3.88	
714E 85	1.5	74.5	6.5	3.0	14.5	0.15	2.29	7.26	31.35	2.02	5.18	0.36	3.59	
714E 90	5.0	67.0	5.5	8.5	14.0	0.0	1.86	7.73	31.13	2.14	5.02	0.42	3.93	
714E 95	0.5	44.5	41.0	2.0	12.0	0.30	1.69	7.66	31.34	2.27	4.47	0.46	3.86	
714E 100	2.0	60.5	2.5	2.5	32.5	0.36	1.06	7.43	33.15	2.07	3.17	0.44	3.57	
714E 105	3.0	62.5	4.0	3.5	27.0	0.0	0.77	7.02	34.23	2.02	3.01	0.45	3.50	
714E 110	4.5	51.5	8.5	5.0	30.5	0.0	1.10	7.57	32.83	2.07	3.36	0.47	3.92	
714E 115	13.5	52.0	4.0	3.5	27.0	0.0	0.80	8.45	32.58	2.11	2.56	0.50	4.38	
714E 120	4.0	65.0	5.5	4.0	21.5	0.11	0.89	8.42	32.52	2.01	2.66	0.54	3.96	
714E 125	4.5	62.5	4.0	2.5	26.5	0.27	0.97	8.46	32.44	1.97	2.81	0.48	4.01	
714E 130	5.0	52.0	8.0	6.5	28.5	0.0	0.97	8.48	32.26	1.93	3.23	0.45	4.20	
714E 135	2.5	66.5	8.5	0.5	22.0	0.15	0.93	8.39	32.76	1.92	2.70	0.47	3.95	
714E 140	8.5	68.0	1.5	5.5	16.5	0.0	0.97	8.76	32.07	1.93	3.00	0.51	4.26	





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
714E 145	9.5	59.0	5.0	4.0	22.5	0.13	0.90	8.77	32.29	1.99	2.79	0.54	3.99
714E 150	4.9	68.6	2.9	5.9	17.6	0.0	0.71	8.56	33.01	1.86	2.61	0.39	3.93
714E 155	7.0	55.5	3.0	7.5	27.0	0.34	0.82	8.43	32.52	2.00	2.85	0.48	3.97
714E 160	5.5	55.0	3.0	2.5	34.0	0.0	0.90	8.31	32.80	1.92	2.84	0.43	4.08
714E 165	1.1	63.8	4.3	3.2	27.7	0.0	0.98	8.27	32.63	2.15	2.90	0.46	4.00
714E 170	16.5	49.0	3.0	0.5	31.0	0.0	0.68	8.38	32.96	2.03	2.25	0.55	4.35
714E 175	6.7	58.1	3.8	3.8	27.6	0.0	0.80	8.32	33.30	1.98	2.01	0.46	4.03
714E 180	9.0	65.0	10.0	2.5	13.5	0.20	0.81	8.48	32.53	1.97	2.59	0.51	4.20
714E 185	6.0	67.5	2.0	1.0	23.5	0.0	0.70	7.93	33.69	1.98	2.06	0.62	3.78
714E 190	2.5	58.5	2.5	1.5	35.0	0.0	0.62	8.19	33.57	1.91	2.22	0.51	3.94
714E 195	1.5	56.5	11.0	0.0	31.0	0.0	0.67	8.17	33.50	1.94	2.26	0.59	3.84
714E 200	2.5	57.0	1.0	2.0	37.5	0.0	0.60	8.43	33.35	1.93	1.93	0.44	4.32
715E 10	4.0	70.0	2.5	3.0	20.5	0.0	1.01	7.65	33.35	2.08	2.63	0.44	4.09
715E 20	2.5	66.5	2.5	5.5	23.0	0.27	0.83	7.39	33.33	2.05	3.22	0.48	3.80
715E 30	2.0	54.5	3.0	4.5	36.0	0.0	0.86	7.00	34.32	2.04	2.75	0.45	3.30
715E 40	2.0	56.5	8.5	5.5	27.5	0.0	0.89	7.59	33.48	1.82	3.33	0.47	3.56
716E 10	2.5	69.0	5.5	5.0	18.0	0.0	0.91	7.53	33.30	1.95	3.32	0.38	3.92
716E 20	26.5	54.0	2.5	2.5	14.5	0.0	0.79	7.78	33.57	2.08	2.14	0.56	4.14
716E 30	48.5	32.5	3.0	1.5	14.5	0.0	1.04	7.79	33.42	2.19	2.58	0.44	3.69
716E 40	1.5	65.5	2.0	4.0	27.0	0.0	0.91	7.55	33.49	2.03	3.12	0.44	3.58
716E 50	1.5	67.0	6.0	3.5	22.0	0.0	1.12	7.62	33.17	2.05	3.01	0.44	3.76
716E 60	2.5	73.0	3.0	2.0	19.5	0.31	1.06	7.52	33.20	2.01	3.07	0.41	3.76
716E 70	0.5	62.5	5.0	5.0	27.0	0.28	1.18	7.50	33.06	2.06	3.12	0.41	3.80



Hole/Depth	CaCO <sub>3</sub>	Cumulative - Percentages - - - - -											Sand, Silt & Clay			
	Equiv.	Sand (Sieve intervals) - - - - -											%Sand	%Silt	%Clay	
		10-20	10-40	10-60	10-140	10-230		%Sand + Silt	%Sand, Silt & Clay	%Coarser than 10 mesh						
7162 50	7.4	3.33	9.89	22.43	36.30	42.34		91.56	100.00	1.49		42.34	49.22		8.44	
7162 60	6.6	3.63	9.89	19.46	34.46	40.52		90.50	100.00	2.12		40.52	49.99		9.50	
7162 70	7.4	3.36	11.58	23.06	36.86	42.81		92.49	100.00	1.59		42.81	49.68		7.51	
7162 80	6.5	3.67	9.80	19.36	34.28	40.00		85.25	100.00	1.84		40.00	45.26		14.75	
7162 90	6.6	3.73	10.05	19.95	33.74	39.79		88.09	100.00	1.10		39.79	48.29		11.91	
7162 100	7.7	4.11	11.56	22.06	38.73	45.59		89.17	100.00	2.58		45.59	43.58		10.83	
7162 110	7.7	1.78	5.84	18.33	41.53	48.08		88.41	100.00	0.92		48.08	40.32		11.59	
7162 120	5.2	2.97	8.15	14.96	28.19	30.86		83.08	100.00	2.34		30.86	52.22		16.92	
7162 130	2.6	0.79	2.55	4.61	7.58	9.62		69.28	100.00	0.40		9.62	59.66		30.72	
7162 140	4.8	3.03	8.42	15.99	29.11	35.30		90.67	100.00	0.30		35.30	55.38		9.33	
7162 150	5.5	3.76	10.39	19.49	32.26	38.13		89.11	100.00	5.54		38.13	50.97		10.89	
7162 160	7.3	3.69	10.26	21.03	35.36	41.61		87.52	100.00	3.09		41.61	45.91		12.48	
7162 170	5.8	5.24	13.51	22.35	38.67	44.89		87.97	100.00	2.52		44.89	43.09		12.03	
7162 180	6.6	4.27	11.69	21.57	36.77	45.73		89.58	100.00	3.80		45.73	43.84		10.42	
7162 190	5.8	3.36	9.38	17.94	32.24	36.95		85.59	100.00	0.69		36.95	48.64		14.41	
7162 200	6.2	3.27	9.22	17.31	32.90	38.55		82.32	100.00	1.62		38.55	43.77		17.68	
7162 210	10.6	3.81	9.86	18.17	32.46	38.40		89.71	100.00	4.21		38.40	51.31		10.29	
7162 220	9.4	2.90	8.01	15.40	28.82	34.32		85.51	100.00	3.05		34.32	51.18		14.49	
7162 230	8.1	3.21	9.73	22.18	37.82	43.30		85.10	100.00	0.68		43.30	41.80		14.90	
7162 240	2.7	2.96	9.26	19.39	31.25	35.61		83.91	100.00	0.88		35.61	48.30		16.09	
7162 250	3.7	3.59	9.78	22.02	33.00	38.50		88.89	100.00	0.50		38.50	50.39		11.11	
7162 260	4.4	4.65	12.11	21.32	35.16	38.82		90.19	100.00	0.90		38.82	51.37		9.81	
7162 270	3.9	4.30	10.71	21.17	34.03	39.65		91.36	100.00	0.85		39.65	51.71		8.64	
7162 280	4.6	4.29	11.96	22.02	36.61	43.05		89.81	100.00	2.11		43.05	46.75		10.19	
7162 290	4.4	4.63	11.51	20.94	34.42	40.08		88.79	100.00	3.25		40.08	48.71		11.21	



Core/Depth Grain Lithology of 1-2mm Fraction (%) Bulk Chemistry of Silt and Clay Fraction (%)

	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
717E 15	1.0	72.0	4.5	2.5	20.0	0.17	0.89	7.25	33.90	2.04	2.31	0.44	4.10
717E 20	1.0	73.0	3.5	1.0	21.5	0.27	0.87	8.62	31.48	2.01	3.61	0.54	4.49
717E 25	1.0	80.0	2.5	1.5	15.0	0.0	0.95	7.24	33.65	2.01	3.50	0.39	3.33
717E 30	4.5	68.5	5.0	2.5	19.5	0.0	1.06	7.36	33.99	1.97	2.82	0.34	3.41
717E 35	1.5	65.8	4.0	5.4	23.3	0.29	1.01	7.35	33.70	2.00	3.27	0.31	3.25
717E 40	3.0	68.0	4.0	1.5	23.5	0.50	1.02	7.18	33.54	2.06	3.05	0.51	3.44
717E 60	3.5	61.0	4.5	4.5	26.5	0.35	1.09	6.51	34.52	1.94	3.01	0.42	3.14
717E 120	14.0	50.5	3.0	4.5	28.0	0.0	0.71	7.87	33.19	1.97	2.44	0.47	4.43
717E 130	9.0	68.5	3.0	2.0	17.5	0.20	0.77	7.93	33.57	2.03	2.06	0.45	4.05
717E 140	5.1	63.4	1.7	2.3	27.4	0.28	0.86	7.59	33.82	1.87	2.69	0.46	3.42
717E 150	2.0	63.0	2.0	4.5	28.5	0.0	0.53	7.36	34.18	2.01	2.63	0.47	3.77
717E 160	7.5	58.0	0.5	0.5	33.5	0.60	0.92	8.05	32.16	2.03	3.69	0.52	3.76
717E 170	3.5	57.5	2.0	7.0	30.0	0.0	0.67	8.01	33.53	1.81	2.29	0.49	4.21
717E 210	5.0	52.0	3.0	12.5	27.5	0.20	0.79	8.46	32.88	1.86	2.32	0.38	4.33
718E 10	4.2	66.3	1.1	0.0	28.4	0.13	0.54	7.91	33.94	2.04	1.67	0.53	4.14
718E 20	3.5	62.0	5.0	5.5	24.0	0.0	0.53	7.19	35.12	1.88	1.82	0.50	3.46
718E 30	63.0	26.0	0.0	0.5	10.5	0.52	0.44	9.52	32.17	1.48	1.59	0.54	4.97
718E 40	3.5	59.0	3.0	6.5	28.0	0.0	0.98	7.19	33.73	2.09	3.08	0.41	3.59
718E 50	2.0	69.5	5.0	1.5	22.0	0.0	0.26	7.85	33.83	1.98	1.85	0.44	4.84
718E 60	7.0	55.5	5.0	4.5	28.0	0.0	0.65	8.13	33.29	2.28	2.34	0.49	3.99
718E 70	3.5	65.5	3.5	5.5	22.0	0.0	0.70	8.28	33.49	2.11	2.10	0.47	3.84
718E 80	12.0	59.5	5.0	6.0	17.5	0.39	0.73	8.11	33.29	1.91	2.28	0.47	3.96
719E 90	4.0	69.5	2.5	4.0	20.0	0.0	0.71	7.92	33.83	1.97	2.29	0.45	3.73



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)				Bulk Chemistry of silt and Clay Fraction (%)								
	Local Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe	
718E 100	5.0	59.5	1.5	6.0	28.0	0.0	0.39	8.26	33.83	1.97	2.02	0.50	3.89
718E 110	3.0	64.0	5.0	2.0	26.0	0.0	0.51	7.26	35.01	1.88	1.80	0.45	3.65
718E 120	10.0	55.5	2.5	3.5	28.5	0.0	0.50	7.75	33.98	2.08	2.05	0.50	4.08
718E 130	11.0	58.5	3.5	0.5	26.5	0.0	0.49	6.90	35.31	1.83	1.79	0.53	3.65
718E 140	4.0	69.5	2.5	2.0	22.0	0.0	0.39	6.96	35.50	1.74	1.48	0.51	3.56
718E 150	7.3	70.9	2.0	3.3	16.6	0.0	0.33	7.88	34.19	2.00	1.84	0.44	4.14
718E 160	10.0	58.5	2.0	7.0	22.5	0.0	0.80	8.38	33.09	1.95	2.36	0.54	3.99
718E 170	6.5	61.5	4.0	1.0	27.0	0.0	0.96	7.49	33.07	2.07	3.66	0.39	3.80
718E 180	7.5	60.5	2.5	5.5	24.0	0.0	0.45	8.35	32.85	2.10	2.56	0.47	4.60
718E 190	4.0	67.3	3.5	2.5	22.8	0.0	0.80	7.88	33.34	2.15	2.47	0.46	4.10
718E 200	1.0	68.5	5.5	2.5	22.5	0.0	0.49	7.67	33.99	2.00	2.24	0.43	4.16
718E 210	2.0	70.5	3.0	5.0	19.5	0.0	0.58	7.40	33.99	2.03	2.36	0.50	4.08
719E 10	8.5	66.0	3.0	4.5	18.0	0.0	0.83	7.54	33.73	2.02	2.77	0.46	3.74
719E 20	7.0	62.5	4.0	2.5	24.0	0.0	1.12	8.05	32.64	2.20	3.30	0.41	3.74
719E 30	4.5	63.0	2.5	4.0	26.0	0.35	1.02	8.12	32.84	2.10	2.87	0.41	3.54
719E 40	1.5	73.5	3.5	2.0	19.5	0.57	1.17	7.23	32.91	2.18	3.64	0.38	3.55
719E 50	0.0	69.5	6.0	4.0	20.5	0.0	1.04	7.46	33.63	2.00	2.79	0.43	3.76
719E 60	1.0	68.5	3.0	1.0	26.5	0.0	0.22	7.53	35.02	1.91	1.84	0.41	3.58
719E 70	3.0	67.5	2.5	4.0	23.0	0.0	0.60	8.76	32.85	2.03	2.07	0.38	4.51
719E 80	3.0	65.5	4.5	1.5	25.5	0.15	0.79	7.77	33.68	1.96	2.13	0.50	3.88
719E 90	1.0	58.5	3.0	7.0	30.5	0.0	0.55	7.22	34.40	1.99	2.44	0.52	3.74
719E 100	1.5	61.0	1.0	5.0	31.5	0.0	0.41	6.91	35.16	1.94	2.12	0.51	3.53
719E 110	6.0	54.5	6.0	5.5	28.0	0.0	0.56	8.58	32.96	2.03	2.24	0.52	4.27





Hole/Depth Grain Lithology of 1-2mm Fraction (%) Bulk Chemistry of Silt and Clay Fraction (%)

	Local	Acid	Basic	Carb.	Otz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
723E 5	4.1	65.5	1.4	0.0	29.1	0.0	0.50	7.51	34.58	2.08	1.38	0.57	4.07
723E 10	2.0	67.0	2.5	4.1	24.4	0.0	0.46	6.98	34.66	1.88	2.38	0.48	3.99
723E 15	8.5	65.5	0.5	4.0	21.5	0.0	0.63	6.95	35.06	1.84	2.32	0.48	3.31
723E 20	1.0	72.5	6.0	1.5	19.0	0.0	0.69	7.23	34.48	1.90	2.45	0.45	3.60
723E 25	2.5	70.5	5.5	1.5	20.0	0.0	0.58	7.35	34.29	1.98	2.39	0.48	3.82
723E 30	5.0	60.0	7.5	4.0	23.5	0.0	0.72	7.38	34.13	1.94	2.44	0.46	3.74
723E 35	3.0	69.0	3.5	3.0	21.5	0.0	0.48	7.16	34.67	2.01	2.35	0.49	3.61
723E 40	7.5	62.5	5.0	3.5	21.5	0.0	0.84	7.20	34.30	1.84	2.51	0.50	3.68
723E 45	7.5	60.0	2.5	5.5	24.5	0.0	0.76	7.30	34.30	1.90	2.39	0.44	3.64
723E 50	6.5	60.5	4.5	2.5	26.0	0.0	0.71	8.20	32.77	2.11	2.84	0.43	4.34
723E 55	1.5	76.0	2.0	1.0	19.5	0.0	0.69	7.03	35.13	1.78	2.33	0.41	3.14
723E 60	5.5	59.0	4.5	2.0	29.0	0.0	0.81	7.61	33.43	1.68	3.26	0.38	4.02
723E 65	4.5	65.0	5.5	1.5	23.5	0.16	0.77	7.42	34.13	1.83	2.58	0.47	3.54
723E 70	7.5	62.0	2.5	2.0	26.0	0.0	0.49	7.49	34.78	1.86	2.31	0.38	3.29
723E 75	10.5	65.0	4.5	2.5	17.5	0.16	0.60	7.08	34.05	1.80	2.32	0.43	3.43
723E 80	8.0	57.5	3.0	2.0	29.5	0.37	0.92	7.27	33.44	1.89	3.71	0.40	3.32
723E 85	18.5	52.0	1.0	2.5	26.0	0.0	0.45	7.46	33.80	1.89	2.84	0.36	4.36
723E 90	5.5	53.5	5.0	2.0	34.0	0.0	0.74	8.04	33.25	1.89	2.79	0.41	4.05
723E 100	4.6	65.6	3.3	1.3	25.2	0.0	0.95	8.41	32.29	1.61	3.39	0.47	4.34
723E 105	34.0	41.5	1.5	1.5	21.5	0.0	0.33	7.94	33.72	1.74	2.78	0.43	4.08
725E 10	4.0	69.0	1.0	1.5	24.5	0.0	1.03	7.62	32.87	2.07	2.54	0.39	4.74
725E 15	6.7	65.9	3.7	2.4	21.3	0.16	1.08	7.33	33.47	2.12	3.05	0.49	3.55
725E 20	2.5	76.0	4.0	1.0	16.5	0.0	1.05	7.53	33.34	2.11	2.83	0.37	4.06
725E 25	1.0	66.5	6.0	0.5	26.0	0.0	1.13	7.63	32.96	2.14	3.15	0.45	3.97



Hole/Depth Grain Lithology of 1-2mm Fraction (%) Bulk Chemistry of Silt and Clay Fraction (%)

Hole/Depth	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
725E 30	16.5	61.5	4.5	1.5	16.0	0.0	1.14	7.85	32.68	1.89	3.43	0.44	4.02
725E 35	11.0	68.5	2.5	1.0	17.0	0.0	1.26	7.76	32.67	1.99	3.38	0.41	3.91
725E 40	7.0	72.5	4.0	2.5	14.0	0.0	0.98	7.54	33.55	2.00	3.12	0.35	3.62
725E 45	2.1	71.3	4.9	1.4	20.3	0.0	1.09	6.85	33.77	2.09	3.53	0.44	3.34
725E 60	8.5	64.5	5.5	3.0	19.5	0.53	1.10	8.22	32.02	1.97	3.25	0.45	4.19
725E 65	45.5	37.0	4.0	0.5	13.0	0.0	1.02	7.82	33.21	1.98	2.73	0.42	4.04
725E 70	41.5	41.0	3.0	1.0	13.5	0.0	0.82	8.29	33.02	2.05	2.45	0.37	4.21
725E 75	1.5	62.0	5.0	3.0	28.5	0.0	1.27	7.76	32.62	2.05	3.44	0.45	3.84
725E 80	1.6	69.2	6.6	0.5	22.0	0.0	1.15	7.55	32.95	1.90	3.65	0.44	3.80
725E 85	4.5	80.3	3.8	0.8	10.6	0.0	1.32	7.53	32.59	1.88	3.76	0.51	3.79
725E 90	13.5	63.5	2.7	0.7	19.6	0.0	0.85	8.76	32.16	2.33	2.51	0.40	4.53
725E 95	50.5	34.0	1.0	7.5	7.0	0.0	0.70	6.23	34.84	1.81	3.60	0.46	3.28
725E 100	17.0	49.5	7.0	4.5	22.0	0.0	0.46	6.50	34.89	1.88	3.38	0.45	3.11
725E 105	12.5	62.5	11.0	2.0	12.0	0.0	0.63	7.67	33.53	2.15	3.09	0.34	3.82
725E 108	6.0	75.0	3.0	2.5	13.5	0.0	1.17	8.18	32.64	2.19	2.86	0.41	3.95
725E 115	34.0	50.0	1.5	2.0	12.5	0.0	0.58	7.35	34.35	2.05	2.85	0.36	3.35
725E 120	14.5	62.0	2.5	3.5	17.5	0.0	0.74	8.04	33.15	2.21	3.01	0.39	3.75
725E 125	41.0	41.5	1.5	2.5	13.5	0.0	1.19	8.14	32.53	2.24	3.15	0.44	3.78
725E 130	20.5	56.0	5.0	2.0	16.5	0.0	0.98	7.40	33.54	2.18	2.86	0.41	3.86
725E 136	2.0	78.5	1.0	2.0	16.5	0.23	1.06	7.59	32.96	2.16	2.93	0.48	3.77
725E 140	13.0	66.5	1.5	2.0	17.0	0.0	0.92	7.75	33.13	2.41	3.28	0.39	3.67
725E 145	6.0	68.7	2.4	3.6	19.3	0.0	0.85	8.30	33.17	2.26	2.40	0.48	3.68
725E 150	1.0	70.5	5.0	4.0	19.5	0.0	1.20	7.81	32.42	2.28	3.54	0.41	4.01
725E 155	10.0	67.5	5.5	1.0	16.0	0.0	0.99	7.69	33.34	2.14	3.11	0.31	3.79
725E 160	6.5	57.5	5.5	2.5	28.0	0.23	1.20	8.33	32.34	2.04	2.91	0.48	3.94



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
725E 165	6.5	68.0	3.5	2.5	19.5	0.20	1.11	7.81	32.82	2.22	3.08	0.39	3.82
725E 170	1.0	66.0	5.5	7.0	20.5	0.0	0.88	7.34	33.63	2.16	3.38	0.40	3.43
725E 175	21.5	59.5	3.0	1.0	15.0	0.0	0.90	7.75	33.63	2.14	2.57	0.32	3.55
725E 180	20.5	54.5	1.5	1.0	22.5	0.0	1.00	9.02	32.02	2.25	2.49	0.47	4.22
725E 190	22.5	59.0	1.0	2.0	15.5	0.0	1.06	7.91	32.92	2.22	2.98	0.38	3.93
725E 195	1.0	73.5	4.0	3.0	18.5	0.0	0.88	7.27	34.04	1.98	3.03	0.37	3.43
725E 200	4.5	64.0	3.5	1.5	26.5	0.0	1.15	7.81	33.08	2.15	3.07	0.41	3.63
725E 210	2.0	75.0	4.5	2.5	16.0	0.0	1.22	7.32	32.98	2.13	3.47	0.35	4.04
725E 220	0.0	68.0	6.0	0.5	25.5	0.0	0.94	7.57	33.20	2.10	3.37	0.48	3.67
725E 230	5.5	63.0	9.5	0.5	21.5	0.0	0.79	8.02	32.99	2.20	3.07	0.27	4.03
725E 240	2.1	68.1	2.1	3.2	24.5	0.0	1.28	8.13	32.28	2.21	3.18	0.46	4.05
725E 250	7.5	63.0	4.0	2.0	23.5	0.0	0.93	8.18	32.04	2.24	3.11	0.50	4.15
725E 255	7.0	64.5	4.5	3.0	21.0	0.0	1.04	7.46	33.38	1.95	3.31	0.39	3.72
726E 10	1.9	71.8	2.9	2.9	20.4	0.0	1.06	8.69	32.33	2.08	2.70	0.40	4.14
726E 20	1.5	73.5	7.0	2.5	15.5	0.0	1.23	8.25	32.25	2.31	3.23	0.34	3.99
726E 30	27.5	53.5	2.0	0.5	16.5	0.0	0.85	7.85	33.34	2.20	2.62	0.37	3.98
726E 40	15.8	61.9	2.2	2.2	18.0	0.16	1.21	7.79	32.95	2.16	2.69	0.38	3.88
726E 50	18.5	55.5	1.5	3.0	21.5	0.59	1.08	7.92	32.50	2.30	2.93	0.39	3.91
726E 55	11.0	61.5	3.5	1.0	23.0	0.24	0.80	7.77	33.73	1.94	2.21	0.43	3.89
726E 60	8.5	55.0	5.0	1.0	30.5	0.46	0.99	7.77	33.24	2.07	2.42	0.47	3.82
726E 70	3.5	58.5	4.0	4.5	29.5	0.0	0.64	7.75	33.90	1.94	2.12	0.48	4.10
726E 80	4.0	63.5	10.0	2.0	20.5	0.0	0.75	7.07	34.26	1.97	2.77	0.50	3.64
726E 90	6.0	59.0	5.0	1.5	28.5	0.25	0.80	7.80	33.48	2.07	2.38	0.45	3.81
726E 100	8.5	52.0	0.5	4.0	35.0	0.0	0.26	7.62	34.08	1.97	2.04	0.49	3.88



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
726E 120	13.0	48.4	4.3	3.1	31.1	0.37	0.72	7.89	33.62	2.10	1.93	0.32	4.14
726E 140	3.5	60.9	2.0	3.0	30.7	0.0	0.42	8.04	34.09	1.89	1.97	0.44	3.92
726E 150	8.5	54.5	3.5	3.5	30.0	0.0	0.69	7.41	34.14	1.94	2.29	0.46	4.00
726E 160	9.5	61.0	3.5	2.5	23.5	0.35	0.54	6.97	34.92	1.99	1.74	0.49	3.69
726E 170	4.2	66.4	4.9	1.4	23.1	0.16	0.46	7.39	34.80	1.99	1.59	0.50	3.75
726E 180	0.0	57.0	4.5	4.0	34.5	0.19	0.50	7.26	34.67	2.00	1.83	0.52	3.77
726E 190	2.5	70.0	6.0	1.0	20.5	0.0	0.55	8.12	33.43	2.00	2.70	0.44	3.85
728E 20	83.0	11.5	0.0	0.0	5.5	0.0	0.76	6.78	34.46	1.85	3.04	0.43	3.64
728E 30	37.6	38.6	4.0	2.5	17.3	0.13	0.96	7.07	33.27	1.74	4.24	0.37	3.44
728E 40	2.5	76.5	4.0	1.5	15.5	0.0	0.80	7.20	34.28	1.84	2.69	0.37	3.75
728E 50	2.0	68.0	6.0	4.5	19.5	0.40	1.48	7.02	32.87	2.03	3.74	0.33	3.77
728E 60	1.0	58.0	4.5	3.0	33.5	0.0	0.48	7.16	34.54	1.62	2.55	0.49	3.95
728E 70	1.0	48.0	30.0	2.5	18.5	0.0	0.50	7.45	34.02	2.01	2.34	0.46	4.25
723E 80	3.5	67.0	0.5	1.0	28.0	0.0	0.53	7.44	34.34	1.90	2.29	0.49	3.97
729E 88	2.5	56.5	8.5	2.0	30.5	0.0	0.73	7.79	33.42	1.74	2.56	0.50	4.40
729E 10	6.5	59.5	4.5	3.5	26.0	0.20	0.83	6.00	35.35	1.63	3.27	0.43	3.00
729E 20	0.5	66.0	5.0	4.0	24.5	0.0	0.83	7.09	34.07	1.82	2.97	0.43	3.82
729E 30	0.5	67.5	5.5	5.0	21.5	0.0	0.82	7.43	33.72	1.90	2.89	0.44	3.92
756E 5	4.4	62.6	3.4	3.4	26.1	0.38	1.29	7.45	32.85	1.98	3.09	0.47	3.93
756E 15	2.0	69.5	3.5	2.0	23.0	0.49	1.51	7.39	32.06	2.09	4.04	0.39	3.97
756E 29	1.5	66.0	4.0	1.5	27.0	0.42	1.17	7.53	33.15	2.03	2.95	0.46	3.64
756E 40	2.0	67.7	2.0	7.0	21.4	0.48	1.02	7.41	33.22	2.02	3.13	0.40	3.71





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)								
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe	
756E 50	3.5	62.5	3.5	4.0	26.5	0.28	1.24	7.48	33.26	2.09	2.90	0.46	3.58	
756E 60	2.0	64.0	5.0	5.0	24.0	0.33	1.05	7.49	33.45	2.17	2.72	0.43	3.59	
756E 70	2.0	72.0	3.5	1.5	21.0	0.32	1.15	7.40	33.30	2.14	2.97	0.38	3.67	
756E 75	2.5	70.5	3.5	3.0	20.5	0.13	1.16	7.53	33.21	2.05	2.85	0.43	3.87	
756E 80	1.0	58.5	7.0	1.5	32.0	0.54	1.14	7.20	33.32	2.07	2.86	0.41	3.72	
756E 90	1.0	72.0	4.0	4.0	19.0	0.59	1.14	7.12	33.44	2.07	2.85	0.43	3.72	
756E 100	9.0	60.2	3.3	1.9	25.6	0.18	0.99	6.60	34.23	1.95	3.12	0.41	3.39	
756E 105	2.0	67.5	5.5	2.0	23.0	0.16	1.16	7.27	33.26	2.13	3.05	0.42	3.85	
756E 110	9.0	65.5	5.0	2.0	18.5	0.31	1.31	7.54	32.69	2.10	3.06	0.50	3.86	
756E 120	1.0	68.5	6.5	3.5	20.5	0.47	1.57	7.67	31.92	2.15	3.96	0.37	3.80	
756E 130	1.5	68.0	6.5	3.0	21.0	0.51	1.55	7.43	32.05	2.17	3.79	0.42	3.92	
756E 135	1.5	77.0	5.0	1.5	15.0	0.38	1.34	8.13	32.56	2.21	2.87	0.41	3.55	
757E 10	1.4	63.1	5.0	4.3	26.2	0.0	0.81	7.40	33.71	2.04	2.62	0.50	3.94	
757E 20	1.0	61.0	3.0	2.0	33.0	0.19	0.98	7.57	32.82	2.04	3.01	0.44	4.33	
757E 30	0.0	66.5	3.5	4.5	25.5	0.21	0.98	7.08	33.69	2.02	3.15	0.37	3.76	
757E 40	3.0	58.0	3.0	7.0	29.0	0.0	0.93	7.20	33.52	2.07	3.21	0.41	3.98	
757E 48	1.0	64.5	2.5	3.0	29.0	0.21	1.00	7.29	33.32	2.11	3.25	0.40	3.82	
757E 50	0.5	62.5	3.5	4.5	29.0	0.0	0.77	7.44	34.17	1.80	2.76	0.41	3.51	
757E 60	0.5	61.0	5.5	3.0	30.0	0.0	1.05	7.51	33.10	2.27	3.19	0.29	4.03	
757E 70	0.0	73.0	5.5	3.5	18.0	0.0	0.88	7.27	33.80	2.06	2.80	0.47	3.85	
757E 80	1.5	68.5	2.5	5.5	22.0	0.0	0.89	7.46	33.30	2.15	3.12	0.46	3.94	
757E 90	1.5	60.5	5.0	2.5	30.5	0.0	1.07	7.22	33.63	2.14	3.14	0.37	3.67	
757E 100	0.5	64.0	5.5	3.0	27.0	0.18	1.08	7.45	33.07	2.14	3.13	0.45	3.94	
757E 108	0.0	65.5	7.0	4.5	23.0	0.0	1.07	7.17	33.65	2.09	2.96	0.44	3.85	



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
757E 320	13.6	53.8	6.0	2.0	24.6	0.0	0.81	8.08	33.01	1.91	2.62	0.51	4.32
758E 60	0.5	42.0	37.5	3.0	17.0	0.48	1.03	7.28	33.35	1.93	3.09	0.44	3.76
758E 68	0.0	65.6	5.3	3.3	25.8	0.36	1.11	7.68	32.78	1.84	3.30	0.45	3.95
759E 90	1.5	68.5	2.5	4.0	23.5	0.0	0.89	7.53	32.88	1.96	3.47	0.50	4.08
759E 120	1.5	70.0	5.0	2.5	21.0	0.0	0.64	7.68	33.51	1.87	2.17	0.46	4.71
758E 150	19.0	60.5	1.9	2.9	15.7	0.0	0.76	7.75	33.26	1.61	2.86	0.62	4.23
758E 180	5.5	62.0	3.0	3.0	26.5	0.0	0.70	7.53	34.13	1.75	2.02	0.54	4.03
758E 210	1.0	44.9	6.1	24.2	23.7	0.0	0.52	7.42	34.19	1.76	2.20	0.52	4.18
758E 240	0.0	62.7	3.0	1.5	32.9	0.0	0.66	7.47	34.00	1.62	2.47	0.58	4.13
753E 270	3.3	68.1	4.8	3.3	20.5	0.0	0.44	7.54	34.45	1.82	1.68	0.50	4.32
758E 300	3.0	53.5	4.5	7.0	32.0	0.0	0.55	7.34	34.65	1.84	1.76	0.43	4.02
758E 330	1.4	71.7	0.7	1.4	24.8	0.0	0.72	8.04	32.80	1.95	2.81	0.56	4.30
759E 20	1.6	65.8	4.1	3.6	24.9	0.0	0.55	7.87	34.01	1.97	2.16	0.38	3.94
759E 30	6.0	67.0	5.5	3.5	18.0	0.0	0.61	7.55	34.24	1.96	2.19	0.40	3.90
759E 43	8.0	59.5	3.0	1.5	28.0	0.0	0.62	7.55	33.74	2.00	2.42	0.52	4.24
759E 50	5.5	59.0	5.0	3.0	27.5	0.0	0.29	7.58	34.31	1.94	2.33	0.57	3.82
760E 10	0.0	61.0	4.0	13.0	22.0	0.81	1.41	7.43	32.04	2.13	3.70	0.42	4.01
760E 20	0.0	70.0	7.0	2.5	20.5	0.66	1.41	7.53	31.56	2.11	4.27	0.51	4.11
760E 30	0.5	71.5	4.5	4.0	19.5	0.35	1.25	7.48	31.85	2.09	4.10	0.49	4.35
760E 40	3.5	58.5	7.5	9.0	21.5	0.54	1.44	7.52	31.96	2.08	4.37	0.47	3.56
760E 50	3.0	79.0	3.0	1.0	14.0	0.72	1.44	7.42	32.00	2.12	4.09	0.45	3.72
760E 60	0.0	68.5	9.5	3.5	18.5	0.68	1.45	6.63	32.50	2.02	4.39	0.67	3.58



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)									
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe		
760E 65	1.5	65.5	7.5	4.0	21.5	0.62	1.21	7.30	32.73	2.07	3.80	0.41	3.53		
760E 70	0.5	67.0	5.0	5.5	22.0	0.67	1.26	7.09	32.66	2.02	4.18	0.39	3.88		
760E 75	7.5	65.0	5.0	2.5	20.0	0.22	0.73	7.33	33.67	1.99	2.44	0.56	4.14		
760E 80	2.5	65.0	6.0	5.5	21.0	0.24	1.31	7.03	32.82	1.96	4.30	0.39	3.55		
760E 90	2.5	67.0	4.5	2.0	24.0	0.23	1.05	7.14	33.25	2.00	3.81	0.44	3.51		
760E 100	6.5	54.5	3.5	4.5	31.0	0.25	0.90	7.08	33.81	1.97	3.44	0.34	3.40		
760E 110	2.0	63.0	3.5	4.0	27.5	0.35	1.31	7.34	32.28	2.13	4.21	0.43	3.81		
760E 120	1.0	62.0	3.5	9.5	24.0	0.58	1.38	7.28	32.55	2.05	3.94	0.43	3.53		
761E 20	1.5	67.3	3.0	4.0	24.3	0.71	1.53	7.47	32.35	2.12	3.34	0.46	3.78		
761E 25	4.2	50.0	23.2	2.6	20.0	0.53	1.10	7.32	33.22	2.12	2.78	0.49	3.83		
761E 40	9.5	70.5	5.0	1.5	13.5	0.49	1.57	7.24	32.27	2.08	4.11	0.40	3.59		
761E 50	1.0	68.5	6.5	4.5	19.5	0.51	1.52	7.10	32.07	2.06	4.03	0.46	4.25		
761E 60	6.0	67.5	3.5	4.0	19.0	0.0	1.27	6.83	32.74	1.97	4.43	0.51	3.82		
761E 70	1.5	63.0	6.5	4.5	24.5	0.51	1.31	7.56	32.31	2.00	3.51	0.48	3.99		
761E 75	3.5	76.5	6.5	3.5	10.0	0.44	1.36	7.60	32.20	2.06	3.87	0.43	3.68		
761E 80	1.0	62.0	3.5	13.0	20.5	0.42	1.49	7.58	31.99	2.04	4.23	0.40	3.74		
761E 90	1.7	76.7	3.3	1.7	16.7	0.70	1.34	6.50	33.27	2.00	3.94	0.44	3.34		
761E 100	0.5	78.0	10.5	1.0	10.0	0.48	1.66	7.14	31.98	2.11	4.30	0.45	3.92		
761E 105	0.0	74.0	3.5	9.0	13.5	0.66	1.55	7.34	31.68	2.02	4.59	0.43	3.73		
761E 115	2.0	71.3	5.9	4.5	16.3	0.54	0.94	7.39	33.10	2.26	3.55	0.39	3.24		
761E 125	2.0	69.5	4.0	4.0	20.5	0.20	1.09	7.38	32.95	2.07	3.67	0.37	3.80		
761E 135	1.0	52.5	5.5	16.0	25.0	0.54	1.19	6.81	33.38	1.97	4.06	0.36	3.09		
761E 150	22.5	49.5	5.5	8.0	14.5	0.20	1.36	7.56	32.27	2.09	3.95	0.43	3.90		



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
762E 20	4.0	75.0	3.4	2.3	15.3	0.21	1.50	7.62	32.45	2.07	3.45	0.44	3.78
762E 240	11.0	62.5	3.5	3.5	19.5	0.32	1.13	7.77	32.55	2.01	3.14	0.43	4.09
762E 245	7.0	58.0	5.0	3.0	27.0	0.58	1.08	7.03	33.38	2.03	3.47	0.46	3.29
763E 40	0.0	76.5	6.0	2.5	15.0	0.75	2.37	6.51	31.20	1.99	5.71	0.46	3.52
763E 50	4.0	58.0	7.5	15.5	15.0	0.66	2.47	6.63	31.08	2.04	5.92	0.40	3.33
763E 60	1.0	74.0	6.0	1.5	17.5	0.56	0.95	7.04	34.45	1.98	3.19	0.34	2.36
763E 75	2.5	70.0	5.5	4.0	18.0	0.27	1.05	7.19	33.03	2.02	3.52	0.42	3.84
763E 80	1.5	57.5	4.5	13.0	23.5	0.27	1.11	6.94	33.68	1.87	3.69	0.36	3.33
763E 90	2.5	70.0	7.5	3.0	17.0	0.49	1.43	7.81	31.94	2.01	4.14	0.46	3.55
763E 100	3.0	67.0	7.5	2.5	20.0	0.38	1.28	7.34	32.61	2.00	3.92	0.41	3.74
763E 110	0.5	57.0	4.0	5.5	33.0	0.37	1.30	7.35	32.34	2.00	4.02	0.47	3.87
764E 5	1.9	66.7	3.3	6.2	21.9	0.50	1.41	7.48	32.32	2.09	3.69	0.48	3.80
764E 45	1.5	57.0	5.0	10.5	26.0	0.56	1.00	9.73	31.80	1.61	3.27	0.35	2.95
764E 55	2.0	70.0	3.5	3.0	21.5	0.25	1.36	7.25	32.62	1.94	4.12	0.47	3.67
764E 65	1.0	63.5	6.5	4.5	24.5	0.52	0.98	7.02	33.46	1.94	3.66	0.45	3.36
764E 90	1.0	59.5	4.0	10.5	25.0	0.47	1.53	6.77	32.81	2.01	4.48	0.38	3.28
764E 98	6.0	63.5	5.0	3.0	22.5	0.73	1.78	7.34	31.86	2.04	4.56	0.41	3.29
764E 111	0.5	75.0	3.5	4.0	17.0	0.50	1.73	7.00	32.10	2.02	4.79	0.40	3.35
764E 120	7.5	55.0	3.0	8.5	26.0	0.66	1.62	7.17	32.15	2.06	4.30	0.38	3.59
764E 125	2.0	75.0	5.5	2.5	15.0	0.47	1.79	6.83	32.48	2.05	4.56	0.40	3.25
764E 135	10.5	65.0	2.5	2.0	20.0	0.48	1.31	7.69	32.69	2.11	3.18	0.42	3.65
764E 144	23.5	43.5	3.5	6.0	23.5	0.55	1.36	7.64	32.47	2.16	3.44	0.36	3.69
764E 155	13.5	56.5	5.5	5.0	19.5	0.13	1.30	7.71	31.22	2.08	5.23	0.40	4.21





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Otz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
765E 30	6.2	64.3	4.3	2.9	22.4	0.39	1.43	7.84	32.40	2.20	3.22	0.45	3.67
765E 50	1.5	61.0	3.5	7.5	26.5	0.35	1.46	7.61	32.60	2.11	3.33	0.41	3.72
765E 70	8.8	54.4	7.4	0.7	28.7	0.12	1.35	8.86	31.53	2.28	2.65	0.44	4.43
765E 90	1.0	73.0	5.5	2.0	18.5	0.58	1.34	7.61	32.35	2.11	3.92	0.41	3.34
765E 110	1.5	63.0	5.0	10.5	20.0	0.39	1.18	7.35	33.20	2.02	3.63	0.37	3.26
765E 297	0.5	63.0	6.0	4.0	26.5	0.0	0.76	7.47	32.61	1.99	2.45	0.45	5.70
765E 300	2.5	69.5	5.5	3.0	19.5	0.28	0.90	7.32	33.49	1.93	3.47	0.43	3.46
795E 20	1.5	55.5	3.5	5.5	34.0	0.27	0.72	7.93	34.06	1.95	2.33	0.40	3.14
795E 25	0.0	73.5	2.0	2.0	22.5	0.35	0.82	7.85	33.85	1.98	2.40	0.34	3.35
795E 50	1.0	68.0	2.5	7.0	21.5	0.37	1.08	6.98	34.00	1.99	2.78	0.45	3.36
795E 75	2.4	64.9	3.4	5.3	22.9	0.36	1.11	7.11	33.77	1.98	3.04	0.39	3.41
795E 145	2.5	65.0	7.5	3.0	22.0	0.24	1.07	7.33	33.63	1.99	2.94	0.39	3.59
797E 20	7.0	71.0	3.0	2.0	17.0	0.0	0.70	7.39	35.22	1.83	2.01	0.31	2.82
797E 40	5.0	62.5	3.5	5.5	23.5	0.0	0.61	8.28	34.51	2.05	1.76	0.37	2.90
797E 60	3.0	62.0	4.0	8.0	23.0	0.38	0.75	7.76	34.38	1.90	2.12	0.35	3.06
797E 80	5.5	58.5	8.0	3.5	24.5	0.09	1.05	7.58	33.69	1.94	2.43	0.48	3.77
797E 100	1.0	67.0	7.5	4.5	20.0	0.09	0.76	7.97	33.95	1.97	2.30	0.42	3.22
797E 170	7.0	59.0	5.0	8.5	20.5	0.64	1.00	8.07	32.59	2.13	2.35	0.45	4.20
797E 190	4.0	64.5	4.0	4.5	23.0	0.18	0.70	7.98	34.18	1.91	1.66	0.44	3.67
797E 230	2.0	64.5	2.0	1.5	30.0	0.0	0.59	7.60	33.94	2.02	2.04	0.57	4.20
797E 250	4.0	62.5	6.5	3.5	23.5	0.14	0.72	8.30	33.58	1.91	1.94	0.41	3.92



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
797E 270	5.0	60.5	3.0	5.5	26.0	0.0	0.66	7.91	33.75	1.93	2.01	0.50	4.04
797E 290	68.5	22.0	0.0	1.5	8.0	0.20	0.79	8.34	32.46	1.98	2.56	0.45	4.73
797E 310	10.0	44.0	4.0	5.5	36.5	0.11	0.56	7.51	34.86	1.80	1.38	0.50	3.66
797E 335	11.5	59.0	4.0	2.5	23.0	0.11	0.64	7.53	34.52	1.85	1.43	0.53	4.05
797E 340	12.2	52.8	9.4	1.1	24.4	0.12	0.73	7.56	34.49	1.85	1.73	0.57	3.59
798E 10	7.5	65.5	5.5	4.0	17.5	0.0	0.56	7.77	34.84	1.63	1.63	0.29	3.71
798E 30	2.6	67.9	5.8	3.2	20.5	0.30	0.92	7.98	33.65	1.98	2.08	0.40	3.65
798E 50	6.0	57.0	7.0	7.0	23.0	0.20	0.93	7.51	34.21	1.90	2.23	0.43	3.40
798E 70	1.0	74.5	5.0	2.5	17.0	0.71	1.95	6.99	32.16	2.10	4.43	0.37	3.21
798E 90	0.5	72.0	5.0	3.5	19.0	0.41	1.94	8.01	31.16	2.18	4.25	0.40	3.64
798E 110	1.0	69.5	7.5	1.0	21.0	0.85	1.21	7.14	33.14	2.03	3.68	0.36	3.09
798E 130	0.5	63.0	11.0	3.5	22.0	0.63	1.23	7.42	32.61	2.14	3.51	0.44	3.65
798E 150	0.0	67.0	7.5	3.0	22.5	0.40	1.48	7.58	32.74	2.19	3.34	0.41	3.37
798E 170	0.0	71.5	5.5	4.0	19.0	0.23	1.54	8.39	31.56	2.49	3.26	0.37	3.98
798E 210	0.0	62.5	3.5	2.0	32.0	0.0	0.58	7.45	35.02	1.93	1.63	0.39	3.47
798E 230	3.0	62.0	3.0	0.5	31.5	0.10	0.54	7.48	34.93	1.85	1.77	0.43	3.40
798E 250	6.0	60.8	2.4	1.8	28.9	0.22	0.77	8.25	33.22	1.99	2.07	0.50	4.09
798E 10	0.0	65.0	5.0	1.5	28.5	0.55	1.34	7.41	33.16	2.03	3.33	0.34	3.20
798E 20	2.0	74.5	2.0	3.5	18.0	0.36	1.45	7.52	32.31	2.13	3.19	0.36	4.35
798E 30	0.0	73.5	4.5	4.0	18.0	0.49	3.02	6.62	29.68	2.02	7.35	0.34	3.67
798E 40	0.5	76.0	7.0	4.5	12.0	0.47	2.42	6.82	31.27	2.03	6.03	0.43	2.87
798E 50	1.0	56.0	3.0	19.0	21.0	0.54	2.67	5.99	31.50	1.94	6.60	0.38	2.80
798E 100	1.5	63.5	6.0	10.0	19.0	0.87	1.32	7.07	32.94	1.95	3.72	0.46	3.23



Hole/Depth Grain Lithology of 1-2mm Fraction (%) Bulk Chemistry of silt and Clay Fraction (%)

Hole/Depth	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
799E 180	2.5	60.0	5.5	3.0	29.0	0.17	0.93	7.82	33.77	2.03	1.83	0.49	3.81
799E 185	0.5	73.0	3.5	1.5	21.5	0.56	1.20	8.19	32.24	2.10	2.98	0.50	3.76
801E 10	0.5	73.5	4.0	4.0	18.0	0.66	1.48	7.05	33.17	2.11	3.31	0.40	3.29
801E 20	0.5	61.0	6.0	2.5	30.0	0.18	1.42	7.35	33.02	2.07	3.53	0.36	3.52
801E 30	2.0	70.0	6.0	4.5	17.5	0.94	1.57	6.75	32.87	2.15	3.69	0.40	3.27
801E 40	21.0	56.5	3.0	1.0	18.5	0.44	1.75	7.75	31.65	2.10	4.07	0.45	3.75
801E 80	57.0	28.5	1.5	1.5	11.5	0.62	1.73	7.90	31.59	2.17	3.93	0.40	3.59
801E 85	6.0	64.5	4.0	2.5	23.0	0.22	1.11	7.39	33.21	2.03	3.20	0.49	3.70
801E 90	1.0	67.5	5.0	4.0	22.5	0.45	1.17	7.40	32.96	1.98	3.20	0.43	3.70
801E 190	8.0	65.5	2.0	2.0	22.5	0.16	0.77	6.92	34.90	1.93	2.24	0.45	3.19
801E 195	1.0	76.5	2.0	3.5	17.0	0.53	0.92	7.51	33.76	2.00	2.28	0.46	3.53
801E 200	3.5	67.0	4.0	3.5	22.0	0.52	0.94	6.60	34.63	1.93	2.77	0.47	2.97
802E 10	2.0	72.5	2.5	0.0	23.0	0.78	1.73	7.68	31.50	2.30	3.73	0.49	3.67
802E 15	2.0	74.0	4.0	1.5	18.5	0.78	1.73	7.68	31.50	2.30	3.73	0.49	3.87
802E 22	2.5	78.0	5.0	1.5	13.0	0.65	2.07	7.25	31.54	2.15	4.67	0.41	3.33
802E 25	3.0	77.0	3.0	1.0	16.0	0.59	1.77	7.14	32.31	2.12	4.15	0.40	3.29
802E 65	0.5	85.5	2.0	4.5	7.5	0.34	4.13	6.01	29.15	1.88	8.65	0.33	2.88
802E 67	0.5	72.0	6.5	4.0	17.0	0.81	3.45	6.88	28.46	1.88	8.08	0.40	3.68
802E 75	1.0	64.5	7.5	3.5	23.5	0.72	2.49	6.16	31.97	2.03	5.66	0.42	2.76
802E 85	3.0	60.0	2.0	2.0	33.0	0.60	2.25	7.49	30.82	2.04	4.94	0.50	3.69
802E 95	4.5	64.0	7.0	1.0	23.5	0.37	0.69	7.42	33.76	1.98	2.64	0.53	3.41
802E 110	3.0	63.5	1.5	0.5	31.5	0.22	1.00	7.16	33.46	1.97	3.40	0.40	3.72
802E 115	3.0	62.5	2.0	5.5	27.0	0.49	1.15	6.90	33.55	1.97	3.53	0.46	3.29



Hole/Depth Grain Lithology of 1-2mm Fraction (%) Bulk Chemistry of Silt and Clay Fraction (%)

	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
802E 140	5.0	60.5	4.5	3.0	27.0	0.17	0.54	6.83	35.64	1.85	1.54	0.46	3.32
802E 175	34.5	31.0	4.5	3.5	26.5	0.22	0.83	8.21	33.02	2.02	2.60	0.45	3.87
802E 185	27.5	43.0	4.5	2.0	23.0	0.18	1.14	8.10	32.37	2.06	3.38	0.51	3.80
802E 195	48.0	29.0	4.0	3.0	16.0	0.0	0.80	8.75	33.02	2.06	1.58	0.48	4.35
804E 10	6.0	68.0	2.5	8.0	15.5	0.23	0.70	6.65	35.35	1.82	1.64	0.39	3.62
804E 20	56.0	25.5	0.5	0.0	18.0	0.56	2.15	6.79	31.85	1.99	5.11	0.40	3.25
804E 25	41.4	35.7	1.4	0.5	21.0	0.33	1.20	4.76	36.83	1.61	3.10	0.36	2.11
804E 30	12.9	61.9	0.5	1.0	23.8	0.46	1.51	6.16	34.54	1.75	3.33	0.43	2.81
804E 55	3.0	65.5	8.0	5.5	18.0	0.54	1.44	5.88	34.67	1.74	3.68	0.40	2.69
804E 60	1.5	66.0	8.5	5.0	19.0	0.39	2.44	6.96	31.11	1.97	5.49	0.42	3.60
804E 65	0.5	66.5	5.0	1.0	27.0	0.65	2.77	6.98	30.32	2.02	6.11	0.39	3.50
804E 70	5.0	67.0	2.5	3.0	22.5	0.78	2.45	6.98	31.01	2.09	5.37	0.37	3.42
804E 75	0.0	72.0	5.0	4.5	18.5	0.39	2.14	7.15	31.68	1.98	4.82	0.42	3.50
804E 80	3.5	66.5	6.0	4.0	20.0	0.23	2.26	8.13	30.23	2.09	5.48	0.37	3.70
804E 95	0.0	65.0	5.0	8.5	21.5	0.66	2.20	7.29	31.10	2.06	5.17	0.42	3.45
804E 100	2.0	64.5	4.5	5.0	24.0	0.55	1.98	7.48	31.47	2.06	4.46	0.42	3.69
804E 105	5.5	60.0	7.0	3.5	24.0	0.53	1.68	7.49	31.98	2.07	4.14	0.38	3.58
804E 108	4.5	54.5	6.0	4.5	30.5	0.35	1.49	7.60	31.43	2.02	4.53	0.50	4.15
804E 110	12.0	71.5	3.0	1.5	12.0	0.61	2.06	6.92	31.74	2.03	4.94	0.34	3.45
804E 115	1.5	59.6	3.9	3.9	31.0	0.60	1.41	7.27	32.31	2.06	4.17	0.42	3.62
804E 120	2.5	64.0	4.5	6.5	22.5	0.37	1.28	7.34	32.63	2.02	3.93	0.48	3.61
804E 125	1.0	58.0	4.5	5.0	31.5	0.33	0.75	6.97	34.74	1.81	2.23	0.43	3.47
804E 130	44.0	46.5	1.0	2.0	6.5	0.40	0.97	8.20	32.44	1.96	2.90	0.64	3.98
804E 185	96.4	2.7	0.0	0.0	0.9	0.0	0.33	10.01	31.83	2.20	1.71	0.50	4.54





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
804E 190	56.5	27.0	0.5	0.0	16.0	0.18	0.87	7.30	34.23	1.87	2.66	0.48	3.29
804E 195	84.5	11.5	0.0	0.0	4.0	0.22	0.77	6.49	35.44	1.79	2.37	0.38	3.08
804E 200	46.8	42.6	2.1	0.0	8.4	0.39	0.83	7.02	34.62	1.83	2.09	0.50	3.49
804E 205	7.0	61.3	6.0	4.0	21.6	0.22	1.03	7.96	32.80	1.96	3.04	0.48	3.88
804E 210	71.0	20.0	1.0	3.0	5.0	0.47	1.04	8.39	32.18	2.01	2.83	0.47	4.09
804E 215	1.0	68.5	6.5	4.0	20.0	0.34	1.00	8.76	31.28	1.98	3.65	0.55	4.35
805E 5	5.5	71.5	1.5	2.0	19.5	0.47	1.65	7.34	32.57	2.03	3.47	0.46	3.64
805E 10	2.5	74.5	2.5	2.5	18.0	0.40	1.50	7.14	33.25	1.99	3.28	0.37	3.45
805E 15	17.6	51.9	2.4	3.8	24.3	0.45	1.37	7.98	32.29	2.01	2.82	0.45	4.23
805E 20	3.0	85.0	2.0	3.5	6.5	0.53	1.76	7.11	31.69	1.86	4.67	0.49	3.83
805E 30	0.0	59.0	5.0	8.0	20.0	0.0	1.97	6.39	32.20	2.08	5.52	0.39	3.35
806E 35	3.5	71.0	3.5	3.5	18.5	0.51	2.20	7.13	31.45	1.94	5.13	0.40	3.43
806E 40	1.0	63.5	9.5	4.5	21.5	0.71	2.10	6.99	31.49	1.89	5.15	0.45	3.46
806E 50	1.0	63.0	5.5	3.5	27.0	0.60	2.21	7.13	31.13	1.97	5.31	0.40	3.54
806E 55	2.5	69.0	3.5	6.5	18.5	0.73	2.21	7.17	31.29	2.01	5.10	0.39	3.38
806E 65	3.5	60.5	8.0	2.0	26.0	0.40	1.30	8.18	31.91	2.04	3.63	0.47	3.79
806E 70	4.0	53.0	7.0	3.0	33.0	0.64	1.28	7.81	32.00	2.15	3.64	0.45	3.88
806E 75	2.0	65.5	3.5	2.5	26.5	0.75	1.09	7.39	32.34	1.92	3.43	0.34	3.68
806E 85	1.0	66.5	4.0	2.5	26.0	0.36	1.15	7.91	32.60	1.93	3.45	0.37	3.70
806E 90	0.0	60.5	7.0	3.5	29.0	0.44	1.20	7.66	32.56	2.05	3.59	0.44	3.67
806E 95	0.0	49.5	5.0	5.5	40.0	0.31	1.08	7.65	32.95	2.01	3.37	0.51	3.54
806E 100	3.5	58.0	4.5	5.0	29.0	0.53	1.17	7.51	32.68	2.04	3.63	0.44	3.55
806E 105	4.0	68.0	3.0	2.5	22.5	0.58	1.10	7.38	32.91	1.87	3.78	0.42	3.48
806E 115	2.5	57.0	3.0	5.5	32.0	0.49	1.32	7.31	32.70	2.07	3.69	0.41	3.51



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
806E 120	5.5	59.5	4.0	6.0	25.0	0.58	1.40	7.53	32.14	2.06	3.86	0.46	3.72
806E 125	3.0	52.0	5.0	4.0	36.0	0.36	1.16	7.69	32.57	2.06	3.49	0.47	3.72
807E 20	9.0	56.5	8.5	3.0	23.0	0.58	1.22	7.44	33.45	1.97	2.70	0.41	3.35
807E 30	62.5	26.0	1.0	1.5	9.0	0.0	1.04	8.04	33.42	1.95	2.45	0.47	3.58
807E 40	76.0	15.0	1.0	2.0	6.0	0.33	1.31	7.84	32.63	2.14	3.04	0.44	3.79
807E 50	4.0	76.5	3.5	1.0	15.0	0.64	1.52	8.13	31.57	2.12	3.87	0.37	3.61
807E 60	15.5	68.0	1.5	2.0	13.0	0.35	1.45	7.98	31.95	2.36	3.26	0.41	3.90
807E 70	25.0	61.5	2.0	0.5	11.0	0.81	1.53	8.28	31.20	2.58	3.53	0.42	3.80
807E 80	24.0	41.5	4.0	2.5	28.0	0.55	1.27	7.83	32.79	1.99	3.04	0.41	3.54
807E 90	24.5	63.5	1.5	0.5	10.0	0.43	1.36	7.56	32.54	2.04	3.44	0.48	3.67
807E 125	7.0	56.0	3.0	5.0	29.0	0.63	1.42	7.93	31.52	2.21	3.95	0.42	3.89
807E 135	7.5	56.0	3.5	4.5	28.5	0.46	1.09	7.97	32.81	2.05	2.84	0.42	3.77
807E 145	1.0	62.0	7.0	1.0	29.0	0.46	1.25	7.94	32.04	2.13	3.39	0.46	4.12
807E 155	2.0	62.0	4.5	3.5	28.0	0.67	1.23	8.19	32.21	2.04	3.03	0.44	3.79
807E 166	0.0	62.5	5.5	3.0	29.0	0.48	0.95	6.11	35.15	1.81	2.78	0.47	3.03
807E 185	6.5	59.0	4.5	2.0	28.0	0.28	0.75	7.88	33.34	1.90	2.48	0.52	3.91
807E 200	4.5	59.5	3.0	4.0	29.0	0.39	0.87	7.23	33.41	2.18	2.56	0.60	4.13
807E 206	2.5	76.5	2.5	2.5	16.0	0.0	1.07	7.77	32.48	1.98	3.87	0.41	4.03
807E 230	2.0	74.5	3.0	2.5	18.0	0.0	0.81	7.36	34.04	1.87	2.82	0.41	3.66
807E 265	16.1	48.7	2.0	7.5	25.6	0.28	1.26	8.73	31.45	2.05	3.31	0.52	4.12
807E 270	3.5	65.5	4.5	3.0	23.5	0.46	0.95	7.76	33.25	1.97	3.09	0.41	3.38
807E 280	6.5	62.0	4.5	6.5	20.5	0.25	1.22	8.44	31.40	2.13	3.64	0.50	4.11
807E 290	5.5	50.5	2.0	4.5	37.5	0.28	0.88	8.84	32.17	1.97	2.73	0.44	4.14
807E 300	5.1	69.1	3.7	2.9	19.1	0.19	1.03	9.14	31.94	2.08	2.59	0.59	3.78



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
807E 310	5.6	70.9	3.9	3.4	16.2	0.10	1.15	8.99	31.08	2.02	3.31	0.49	4.57
809E 5	11.9	48.6	1.9	4.8	32.9	0.20	0.99	7.61	33.88	1.95	1.97	0.50	3.85
803E 25	45.5	34.5	4.0	2.0	14.0	0.33	0.86	7.00	35.09	1.87	1.77	0.36	3.28
808E 35	3.0	77.0	1.0	2.0	17.0	0.40	1.20	8.25	31.80	2.16	3.04	0.56	4.39
808E 45	7.0	72.5	3.5	2.0	15.0	0.74	1.61	7.52	32.18	2.09	3.60	0.43	3.63
808E 55	1.0	65.0	6.5	5.0	22.5	0.69	1.40	7.37	33.18	2.10	3.22	0.37	3.02
808E 65	18.5	56.5	5.0	0.5	19.5	0.0	1.00	7.72	33.34	2.01	2.68	0.46	3.99
803E 75	44.5	31.0	1.5	3.5	19.5	0.0	0.99	7.41	33.93	1.90	2.60	0.51	3.63
808E 85	14.5	64.0	3.5	1.5	16.5	0.0	1.28	7.98	32.59	2.02	2.93	0.42	4.05
808E 95	17.5	44.0	6.5	6.5	25.5	0.33	1.10	7.70	33.13	2.06	2.78	0.48	3.72
809E 105	28.9	35.6	0.7	0.7	34.2	0.0	1.03	7.98	32.79	2.13	2.66	0.51	3.93
808E 125	16.5	49.5	5.0	6.5	22.5	0.39	0.67	8.12	33.73	2.03	1.34	0.40	4.39
809E 134	20.5	60.5	5.0	2.0	12.0	0.43	1.03	7.57	33.19	2.11	2.74	0.43	3.77
808E 185	11.5	52.5	3.5	7.0	25.5	0.36	1.11	7.88	32.75	2.07	3.27	0.49	3.50
808E 195	35.0	50.0	2.0	2.0	11.0	0.33	1.15	7.52	33.40	2.03	2.77	0.45	3.55
808E 206	15.0	59.5	4.0	3.5	18.0	0.25	0.67	7.72	34.46	1.94	1.45	0.42	3.62
808E 215	73.1	14.3	1.7	1.7	9.2	0.07	0.44	6.65	36.72	1.84	0.87	0.39	2.87
808E 320	8.5	51.5	4.0	7.5	28.5	0.96	0.94	7.08	34.16	1.90	2.75	0.39	2.79
809E 325	9.5	65.0	3.0	4.5	18.0	0.46	0.78	8.02	33.74	1.92	1.73	0.45	3.52
808E 330	9.5	65.0	4.5	3.5	17.5	0.37	1.20	8.18	32.09	2.12	3.23	0.50	3.99
803E 335	5.0	69.2	3.0	3.5	19.4	0.31	1.12	7.97	32.54	2.05	3.26	0.42	3.85
808E 340	4.0	52.5	2.0	6.6	34.8	0.64	1.21	8.41	31.10	2.23	3.56	0.54	4.45
809E 15	2.5	51.5	3.5	4.5	38.0	0.0	1.05	8.90	32.14	2.03	2.64	0.43	3.84



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Otz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
809E 25	3.5	68.0	2.0	4.5	22.0	0.28	0.70	7.75	34.42	1.87	1.36	0.48	3.72
809E 35	5.0	44.5	4.0	2.0	44.5	0.30	0.77	8.49	33.21	1.99	1.82	0.48	3.98
809E 45	4.0	69.5	1.0	2.0	23.5	0.20	0.85	7.94	33.91	1.87	1.83	0.48	3.69
809E 55	0.5	73.5	2.5	3.0	20.5	0.45	0.85	7.84	33.53	1.96	1.96	0.49	3.94
809E 65	6.5	68.5	2.0	1.0	22.0	0.48	1.04	7.99	33.00	2.03	2.03	0.47	4.20
809E 75	4.5	50.5	3.0	4.0	38.0	0.32	0.65	8.23	33.80	1.83	1.61	0.49	3.90
809E 85	11.4	69.6	5.1	0.6	13.3	0.15	0.70	7.78	34.57	1.91	1.24	0.57	3.62
809E 95	8.5	46.5	8.0	3.0	34.0	0.40	0.75	7.73	34.09	1.90	1.65	0.50	3.71
809E 105	1.0	79.0	2.0	1.0	17.0	0.22	0.71	7.30	34.67	1.90	1.49	0.51	3.88
809E 115	4.5	58.0	1.5	3.5	32.5	0.19	0.65	7.42	34.60	1.88	1.65	0.49	3.80
828E 10	23.5	49.5	3.0	2.0	22.0	0.0	0.72	8.33	33.53	2.09	1.51	0.49	4.19
828E 20	10.0	56.0	4.5	1.5	28.0	0.37	1.17	7.70	32.99	2.14	2.79	0.47	3.73
828E 30	3.0	66.5	4.5	2.0	24.0	0.60	1.28	7.51	32.60	2.09	3.10	0.45	3.94
828E 40	5.0	65.0	5.0	1.5	23.5	0.50	1.02	7.36	33.48	2.04	2.88	0.43	3.59
828E 50	0.5	69.0	3.0	3.5	24.0	0.58	1.27	7.53	32.63	2.28	2.96	0.47	3.93
828E 60	5.5	72.0	4.0	2.0	16.5	0.29	1.00	7.30	33.19	2.20	3.18	0.44	3.70
828E 70	6.0	61.0	3.5	5.5	24.0	0.29	1.10	6.63	34.20	2.12	3.07	0.35	3.23
829E 5	0.0	76.5	0.5	0.5	22.5	0.72	0.54	7.75	34.15	2.29	1.27	0.40	3.81
829E 10	5.6	70.6	2.1	4.9	16.8	0.51	1.25	6.51	33.99	1.92	3.43	0.30	3.35
829E 15	0.0	75.5	3.0	1.0	20.5	0.81	1.35	7.19	32.50	2.38	3.65	0.37	3.59
829E 20	0.5	83.0	4.0	1.0	11.5	0.51	0.44	7.50	33.59	2.25	2.01	0.72	4.22
829E 25	1.0	50.0	34.5	2.5	12.0	0.23	0.54	7.68	34.49	2.32	1.25	0.43	3.72
829E 30	1.0	81.0	3.5	2.5	12.0	0.39	1.21	6.94	33.23	2.20	3.45	0.33	3.81





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
829E 35	1.5	80.5	5.0	0.5	12.5	0.25	1.30	7.32	33.01	1.97	3.84	0.33	3.45
829E 40	2.0	79.5	4.0	0.0	14.5	0.36	1.22	7.57	32.73	2.06	3.38	0.43	3.80
829E 45	0.0	74.5	7.5	0.5	17.5	0.51	1.49	7.44	32.44	2.22	3.51	0.46	3.65
829E 50	0.0	71.0	7.5	1.0	20.5	0.40	1.29	7.44	32.65	2.17	3.68	0.41	3.59
829E 55	2.0	78.0	4.0	1.5	14.5	0.56	1.90	7.71	30.94	2.28	4.35	0.43	4.20
829E 60	4.0	79.0	5.0	0.0	12.0	0.19	0.62	7.61	34.57	1.87	1.31	0.48	3.99
829E 65	0.5	74.0	3.0	12.0	10.5	0.81	0.30	8.75	32.29	1.78	2.46	0.49	3.92
829E 70	2.5	78.0	2.5	2.5	14.5	0.19	0.75	7.39	34.50	1.91	1.61	0.54	3.93
829E 75	1.5	72.0	5.5	1.5	19.5	0.40	0.62	7.23	35.08	1.89	1.57	0.44	3.69
829E 80	2.5	67.0	6.0	6.0	18.5	0.22	0.81	7.28	34.55	1.82	1.56	0.56	3.90
829E 85	7.5	67.5	2.5	2.5	20.0	0.10	0.57	7.47	34.89	1.84	1.28	0.55	3.80
829E 90	1.5	60.5	14.0	1.5	22.5	0.40	0.31	7.30	35.68	1.72	1.12	0.46	3.61
829E 95	1.5	72.5	3.5	3.5	19.0	0.34	1.60	7.82	31.95	2.20	3.39	0.46	4.07
830E 5	0.8	78.2	5.6	0.8	14.5	0.54	2.58	6.62	30.58	2.06	6.45	0.38	3.57
830E 10	12.1	63.6	7.6	0.8	15.9	0.44	1.15	7.64	33.15	2.03	2.52	0.37	3.92
830E 45	0.0	79.0	2.0	8.5	10.5	0.27	0.92	8.42	32.63	1.91	2.24	0.50	4.44
830E 65	2.0	69.0	3.5	2.5	23.0	0.25	1.93	7.55	31.46	2.20	4.44	0.45	3.84
830E 70	0.5	57.5	9.0	3.0	30.0	0.42	1.47	7.23	31.92	2.10	4.23	0.48	4.00
830E 75	2.0	53.5	23.0	4.0	17.5	0.47	1.29	7.23	32.88	1.99	3.66	0.42	3.64
830E 80	4.5	65.5	6.0	2.0	22.0	0.48	2.07	7.53	31.01	2.19	4.90	0.42	3.74
830E 85	0.0	64.5	8.0	3.5	24.0	0.42	0.95	7.15	34.01	1.96	2.96	0.40	3.13
830E 90	4.0	66.0	5.5	2.0	22.5	0.57	1.21	7.24	32.41	2.05	4.01	0.42	3.93
830E 95	0.5	73.5	5.5	1.5	19.0	0.55	1.18	7.15	32.73	2.05	3.77	0.41	3.69
830E 120	1.5	62.5	9.0	3.5	23.5	0.60	1.08	6.45	33.75	1.94	3.64	0.42	3.36



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)				Bulk Chemistry of Silt and Clay Fraction (%)									
	Local	Acid	Basic	Carb. Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe		
830E 153	3.0	59.3	5.9	7.4	24.4	0.78	1.37	7.00	32.34	2.15	4.01	0.40	3.83	
830E 164	2.5	66.0	6.5	1.0	24.0	0.52	1.14	7.64	33.09	2.10	2.79	0.39	3.71	
831E 10	1.5	66.0	7.0	4.5	21.0	0.55	1.45	7.81	31.92	2.23	3.32	0.41	3.94	
831E 40	2.5	52.0	23.5	5.0	17.0	0.72	1.52	7.05	32.20	2.14	4.23	0.40	3.51	
831E 60	1.5	58.0	4.0	10.0	26.5	0.44	1.33	7.12	33.11	1.94	3.69	0.33	3.31	
831E 80	2.0	56.0	6.5	7.5	28.0	0.50	1.39	6.98	32.81	2.07	4.03	0.40	3.52	
831E 100	2.0	58.0	7.5	4.0	28.5	0.51	1.20	7.30	32.95	2.05	3.43	0.44	3.66	
831E 140	0.5	64.0	12.0	1.5	22.0	0.65	1.02	7.45	32.63	2.15	3.31	0.48	3.88	
831E 160	0.0	71.5	2.5	1.5	24.5	0.25	1.06	7.62	32.93	2.23	3.15	0.41	3.84	
831E 180	1.5	62.5	6.0	3.5	26.5	0.53	1.04	8.22	32.83	2.14	2.57	0.39	3.47	
831E 200	1.5	69.0	3.0	2.0	24.5	0.62	1.26	7.88	32.44	2.21	3.08	0.43	3.70	
831E 220	2.0	55.5	4.0	4.0	34.5	0.77	1.28	7.72	32.44	2.26	3.09	0.42	3.70	
831E 280	4.0	74.0	6.5	3.5	12.0	0.43	0.94	7.75	33.48	2.04	2.17	0.49	3.85	
832E 10	2.5	73.8	3.5	4.5	15.0	0.59	1.35	7.19	33.29	2.09	3.04	0.38	3.39	
832E 20	0.0	75.0	3.0	2.0	20.0	0.52	1.49	7.02	32.88	2.11	3.49	0.42	3.49	
832E 30	13.5	61.5	3.0	1.0	21.0	0.56	1.59	7.80	32.27	2.09	3.27	0.41	3.68	
832E 40	0.0	72.0	4.5	2.0	21.5	0.55	1.74	7.64	32.05	2.14	3.44	0.43	3.63	
832E 50	3.0	61.9	6.5	3.0	25.6	0.52	1.30	8.50	32.10	2.17	2.64	0.51	3.87	
832E 60	1.5	63.0	5.0	3.0	27.5	0.64	1.42	7.64	32.61	2.22	2.94	0.40	3.63	
832E 70	1.0	70.0	8.5	1.5	19.0	0.40	1.19	7.04	34.11	2.07	2.79	0.34	3.00	
832E 80	0.5	76.0	5.0	1.5	17.0	0.37	1.78	7.77	31.32	2.03	3.56	0.48	3.88	
832E 90	1.5	64.0	8.5	3.0	23.0	0.57	1.53	7.07	33.19	2.11	3.38	0.43	3.25	
832E 100	4.2	64.6	5.2	1.0	25.0	0.44	1.09	8.19	33.00	1.99	2.09	0.51	3.89	



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	K <sub>2</sub>	Al	Si	K	Ca	Ti	Fe
832E 110	3.2	69.6	8.0	0.8	18.4	0.65	1.32	6.58	34.25	1.90	2.87	0.44	2.97
832E 120	1.9	74.5	0.9	2.8	19.8	0.61	0.98	7.88	33.20	2.03	2.20	0.52	3.63
832E 130	0.5	72.0	11.0	4.5	12.0	0.56	0.81	7.27	34.52	1.95	1.65	0.49	3.51
832E 140	6.0	71.0	5.5	6.5	11.0	0.32	0.69	7.02	35.48	1.83	1.08	0.42	3.52
832E 150	1.5	72.5	6.0	5.5	14.5	0.44	1.41	7.52	32.88	2.03	3.01	0.43	3.59
832E 160	2.0	60.0	4.5	4.5	29.0	0.35	1.45	7.96	32.16	2.17	3.02	0.46	3.85
832E 170	7.5	54.5	2.5	6.0	29.5	0.95	1.34	8.12	32.65	2.32	2.47	0.39	3.19
832E 180	1.5	68.0	4.0	2.5	24.0	0.18	1.20	8.34	32.35	2.11	2.99	0.41	3.90
832E 190	1.0	57.5	4.5	5.5	31.5	0.43	0.82	8.27	33.15	1.99	2.01	0.50	3.95
832E 198	5.0	71.0	0.5	1.0	22.5	0.46	1.12	8.51	32.16	2.09	2.55	0.49	3.99
833E 10	74.5	17.0	1.5	1.5	5.5	0.26	1.28	7.13	33.35	2.12	3.19	0.39	3.65
833E 20	40.5	45.0	1.0	0.5	13.0	0.18	1.14	7.88	33.15	2.12	2.15	0.46	4.14
833E 30	4.5	63.5	2.0	6.5	23.5	0.0	1.11	7.64	33.53	2.14	2.45	0.51	3.73
833E 40	26.0	54.0	4.0	2.5	13.5	0.25	1.23	7.61	32.75	2.10	3.03	0.49	4.06
833E 50	84.0	10.0	3.0	1.5	1.5	0.28	1.25	7.29	33.19	2.09	3.14	0.41	3.76
833E 60	48.5	33.0	3.0	1.0	14.5	0.44	1.03	7.33	33.70	2.18	2.49	0.42	3.60
833E 70	14.2	58.9	7.1	1.4	18.4	0.27	1.22	7.43	33.15	2.05	2.96	0.41	3.88
833E 80	38.5	46.5	2.0	1.0	12.0	0.45	1.28	7.91	32.34	2.20	3.05	0.38	4.05
833E 90	28.5	45.5	4.5	2.5	19.0	0.0	0.92	7.51	33.96	1.95	2.46	0.40	3.76
833E 100	37.5	44.0	2.5	2.0	14.0	0.0	1.00	7.66	33.58	1.95	2.53	0.50	3.71
833E 115	4.0	59.0	8.0	8.0	21.0	0.12	0.68	7.15	34.91	1.99	1.98	0.37	3.41
833E 125	1.0	69.5	4.0	4.0	21.5	0.39	1.73	7.03	32.50	2.07	4.34	0.45	3.25
833E 135	10.5	63.5	5.0	9.5	12.5	0.61	2.32	7.22	30.89	2.09	5.14	0.43	3.70
833E 200	12.5	59.0	1.0	2.5	25.0	0.51	1.62	7.83	31.39	2.18	4.09	0.43	3.94



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)				Bulk Chemistry of Silt and Clay Fraction (%)									
	Local Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe		
833E 210	22.5	47.0	1.5	1.0	28.0	0.13	0.69	7.76	34.18	2.03	1.56	0.50	3.95	
833E 220	13.0	53.0	3.0	1.0	30.0	0.27	0.69	7.64	33.90	1.99	1.47	0.49	4.47	
833E 230	47.0	31.0	2.0	3.0	17.0	0.15	0.70	7.76	34.16	1.95	1.49	0.44	4.14	
833E 240	5.4	65.9	3.0	4.8	21.0	0.22	0.83	8.03	33.41	2.01	1.83	0.54	4.13	
833E 250	6.0	54.0	3.0	0.5	36.5	0.10	0.96	8.34	32.90	2.02	2.28	0.46	4.03	
833E 260	11.5	53.5	3.5	0.5	31.0	0.38	0.75	8.27	32.89	2.14	2.29	0.49	4.13	
833E 270	29.0	38.5	2.5	6.5	23.5	0.14	0.85	8.40	32.84	1.97	2.38	0.48	4.20	
833E 280	48.0	47.0	0.0	1.0	4.0	0.20	0.69	8.05	33.69	1.92	2.00	0.54	3.85	
834E 5	12.2	55.1	3.1	3.1	26.5	0.0	0.90	7.55	34.22	1.88	2.01	0.46	3.89	
834E 15	3.0	69.0	3.0	1.5	23.5	0.0	0.76	7.55	33.94	1.97	2.35	0.45	3.78	
834E 25	0.5	73.5	4.0	1.0	21.0	0.71	1.41	7.34	32.83	2.10	3.40	0.44	3.32	
834E 35	4.0	58.0	3.0	6.0	29.0	0.21	0.73	6.99	34.67	1.91	2.36	0.41	3.39	
834E 45	1.5	63.0	5.5	4.5	25.5	0.24	1.15	6.82	33.81	1.91	3.34	0.44	3.31	
834E 55	2.5	65.0	5.0	2.5	25.0	0.33	1.30	7.35	32.88	2.04	3.42	0.45	3.66	
834E 65	6.5	53.5	6.0	5.0	29.0	0.48	1.17	7.54	32.54	2.18	3.32	0.44	4.04	
834E 85	5.0	65.5	8.0	2.0	19.5	0.38	1.17	7.56	32.64	2.17	3.45	0.38	3.89	
834E 95	19.0	54.5	4.5	0.5	21.5	0.31	1.10	7.89	32.87	2.05	2.82	0.39	3.86	
834E 160	1.0	59.0	8.5	6.5	25.0	0.29	0.83	7.58	33.30	2.06	2.79	0.53	3.89	
834E 170	7.0	64.0	4.0	2.0	23.0	0.13	0.63	7.46	34.36	1.92	2.18	0.46	3.65	
834E 180	2.0	66.5	1.0	4.5	26.0	0.24	0.72	7.75	33.89	1.93	1.96	0.53	3.91	
834E 190	4.5	64.5	3.5	2.5	25.0	0.59	0.70	7.40	34.14	1.99	1.98	0.46	3.69	
835E 10	3.0	69.0	1.5	1.0	25.5	0.44	1.08	8.14	33.01	2.01	2.00	0.50	4.00	
835E 30	13.5	61.0	3.5	1.5	20.5	0.26	1.11	7.94	32.71	2.03	2.42	0.47	4.20	





Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
835E 40	3.5	59.5	5.0	5.5	26.5	0.38	1.16	8.14	32.91	2.25	2.51	0.37	3.48
835E 50	14.0	51.0	3.5	3.5	28.0	0.44	1.25	8.30	32.60	2.28	2.48	0.37	3.63
835E 60	14.5	46.0	14.0	5.0	20.5	0.33	1.05	8.42	32.03	1.89	2.84	0.51	4.47
835E 70	26.5	43.5	3.5	2.5	24.0	0.72	2.53	7.01	30.85	2.28	5.52	0.38	3.26
835E 80	1.5	73.5	2.0	2.5	20.5	0.43	1.39	6.99	33.75	1.96	2.92	0.46	3.26
835E 90	23.5	46.5	4.0	5.5	20.5	0.39	1.03	7.76	33.32	2.06	2.53	0.41	3.70
835E 100	22.5	57.5	2.5	3.5	14.0	0.38	1.04	7.55	33.68	2.25	2.47	0.40	3.35
835E 150	4.5	53.5	7.0	3.5	31.5	0.50	1.44	6.61	33.81	2.05	3.37	0.33	3.13
835E 160	3.7	65.2	3.7	3.0	24.4	0.65	2.92	6.92	30.20	2.09	6.62	0.33	3.09
835E 170	5.5	58.5	1.0	2.0	33.0	0.85	2.35	6.73	31.58	2.20	5.15	0.37	3.06
835E 180	9.5	56.5	8.5	2.0	23.5	0.36	1.38	7.49	33.06	2.14	2.99	0.41	3.51
836E 30	4.5	55.5	12.5	4.5	23.0	0.56	1.13	9.04	31.64	2.36	2.76	0.45	3.61
836E 60	3.0	55.0	5.5	5.0	31.5	0.21	1.26	7.63	33.29	2.04	2.93	0.39	3.48
836E 100	21.3	45.7	4.7	3.1	25.2	0.41	1.01	8.25	33.01	2.08	2.25	0.41	3.72
836E 140	1.5	56.0	7.5	6.5	28.5	0.43	1.14	7.76	32.89	2.20	2.73	0.39	3.84
836E 200	1.8	61.5	2.8	0.0	33.9	0.44	1.42	8.48	31.71	2.18	3.15	0.45	3.93
952E 10	1.5	80.0	2.0	0.0	16.5	0.31	1.25	8.84	31.77	2.21	2.65	0.45	4.09
952E 20	3.5	65.0	6.0	4.0	21.5	0.42	0.86	8.44	33.39	1.97	1.89	0.44	3.56
952E 24	4.5	71.0	8.0	1.5	15.0	0.24	0.97	7.86	33.82	1.99	1.83	0.43	3.79
952E 30	31.0	44.0	3.0	3.0	19.0	0.51	1.27	8.42	31.91	2.09	2.59	0.51	4.40
952E 40	36.5	37.0	3.5	6.0	17.0	0.25	1.01	7.93	33.32	2.03	1.93	0.51	3.97
952E 65	4.5	72.5	1.5	2.0	19.5	0.44	1.28	7.73	32.71	2.17	3.07	0.51	3.58
952E 70	1.0	67.5	2.5	2.5	26.5	0.46	1.24	7.81	32.87	2.08	2.82	0.42	3.70



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)								
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe	
952E 75	2.0	65.5	7.0	2.0	23.5	0.44	1.36	7.83	32.49	2.13	3.06	0.45	3.64	
952E 80	15.5	62.5	2.5	1.0	18.5	0.39	1.20	7.57	32.96	2.09	3.04	0.44	3.73	
952E 95	1.5	63.0	4.0	3.5	28.0	0.52	1.28	8.37	31.87	2.15	2.79	0.43	4.23	
952E 105	3.5	70.5	4.5	5.0	16.5	0.32	0.64	7.62	34.25	1.88	1.42	0.49	4.08	
952E 110	5.5	57.5	6.5	9.5	21.0	0.45	1.27	7.97	32.45	2.13	2.80	0.44	3.95	
953E 6	3.4	75.7	0.7	0.7	19.6	0.38	0.88	8.16	33.19	1.83	2.11	0.46	4.12	
953E 60	2.0	42.0	9.5	22.5	24.0	0.34	0.79	7.81	33.82	1.96	1.96	0.44	3.84	
953E 70	1.5	65.0	3.0	4.0	26.5	0.47	1.09	8.16	32.80	2.07	2.14	0.45	4.04	
953E 80	0.0	66.0	1.5	2.5	30.0	0.11	0.45	7.26	35.30	1.79	1.46	0.49	3.40	
953E 90	3.5	70.0	2.5	2.5	21.5	0.22	0.73	7.96	33.96	1.97	1.69	0.48	3.81	
953E 100	3.5	34.0	1.5	2.0	59.0	0.0	0.58	7.71	34.44	1.86	1.54	0.47	3.82	
953E 105	2.0	65.3	5.5	3.5	23.6	0.29	1.08	7.98	33.26	2.08	2.57	0.37	3.52	
953E 110	4.5	57.5	2.2	3.0	32.8	0.0	0.75	8.47	33.25	1.89	1.69	0.74	4.10	
953E 120	3.8	61.7	1.5	1.5	31.6	0.0	0.52	7.65	34.65	1.83	1.58	0.51	3.85	
953E 130	5.5	31.0	3.0	1.5	59.0	0.0	0.68	8.26	33.98	1.91	1.54	0.48	3.77	
953E 140	3.7	55.9	2.5	4.3	33.5	0.17	0.62	7.75	34.60	1.95	1.38	0.45	3.66	
953E 151	5.2	57.4	8.7	3.5	25.2	0.42	1.02	8.12	32.98	1.98	2.19	0.56	3.97	
953E 160	20.0	52.9	3.6	3.6	20.0	0.15	0.65	6.96	35.19	1.75	1.84	0.37	3.62	
956E 10	0.0	76.5	5.5	1.0	17.0	0.67	1.37	7.98	32.17	2.23	3.17	0.36	3.78	
956E 20	9.5	56.0	4.5	4.0	26.0	0.23	0.62	8.01	34.21	1.84	1.85	0.41	3.52	
956E 30	10.0	54.0	7.0	4.0	25.0	0.23	0.94	8.46	32.74	1.97	2.32	0.44	4.04	
956E 40	14.0	47.5	3.5	6.5	28.5	0.41	1.18	7.90	32.12	1.92	3.20	0.49	4.39	
956E 50	13.0	55.5	2.0	1.5	28.0	0.23	1.05	7.83	32.93	1.85	2.57	0.51	4.36	



Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
956E 60	6.5	62.0	2.5	4.5	24.5	0.0	1.00	8.31	33.29	1.26	2.30	0.45	4.24
956E 70	42.5	32.5	2.0	2.5	20.5	0.0	0.70	7.06	34.35	1.90	1.88	0.42	4.54
956E 80	9.0	59.2	4.0	2.5	25.4	0.0	0.77	8.05	33.28	1.99	2.15	0.51	4.39
956E 90	3.5	63.0	3.0	5.5	25.0	0.22	0.97	8.38	32.02	1.93	2.73	0.53	4.87
956E 100	3.0	64.0	3.0	3.5	26.5	0.25	0.80	8.11	33.11	1.89	2.44	0.45	4.17
956E 110	9.5	56.0	3.5	4.5	26.5	0.38	1.11	7.58	33.18	2.00	2.54	0.45	3.89
956E 120	10.0	54.5	7.0	3.5	25.0	0.37	1.02	8.37	32.51	1.94	2.56	0.48	4.17
956E 130	5.5	65.0	2.5	3.5	23.5	0.49	1.09	8.11	32.37	1.86	2.69	0.48	4.46
956E 140	9.0	60.5	5.5	5.0	20.0	0.46	1.22	8.60	31.63	1.94	2.85	0.47	4.60
956E 150	7.5	61.0	3.5	3.0	25.0	0.36	1.03	8.65	32.25	1.94	2.40	0.50	4.31
956E 160	16.5	57.5	1.5	4.5	20.0	0.23	0.69	8.25	33.87	1.83	1.75	0.37	3.73
956E 170	6.0	60.3	4.0	6.5	23.1	0.57	2.07	6.62	32.12	1.93	4.98	0.39	3.26
956E 200	4.5	77.5	1.0	1.0	16.0	0.31	0.68	7.29	35.12	1.82	1.44	0.42	3.36
956E 210	8.2	62.8	7.1	3.1	18.9	0.42	0.87	7.57	34.14	1.92	1.80	0.47	3.66
956E 220	5.5	57.0	4.5	3.0	30.0	0.20	0.45	7.69	35.29	1.81	1.13	0.36	3.32
956E 233	9.5	64.0	2.5	4.5	19.5	0.53	0.71	8.35	33.09	1.66	2.06	0.54	4.07
956E 240	10.0	58.0	4.5	5.0	22.5	1.52	0.75	8.07	32.02	1.84	2.66	0.50	4.32
956E 250	13.5	52.0	6.0	3.0	25.5	0.39	0.56	7.49	34.55	1.79	1.38	0.49	3.82
956E 260	8.5	57.0	8.5	1.5	24.5	0.46	0.64	7.90	33.55	1.97	1.78	0.50	4.13
956E 270	3.5	62.0	3.5	0.5	30.5	0.64	0.60	7.60	33.72	1.95	1.65	0.51	3.98
956E 280	11.0	55.5	5.0	3.0	25.5	0.47	0.53	7.61	33.95	1.85	1.66	0.40	4.30
956E 290	11.5	58.0	4.5	3.0	23.0	0.24	0.72	8.08	33.42	1.97	1.79	0.53	4.34
956E 300	13.5	47.5	6.5	3.0	29.5	0.0	0.33	7.09	35.40	1.77	1.40	0.45	3.98
957E 20	3.0	63.0	8.0	6.0	20.0	0.42	1.32	8.05	31.64	2.14	3.16	0.47	4.20



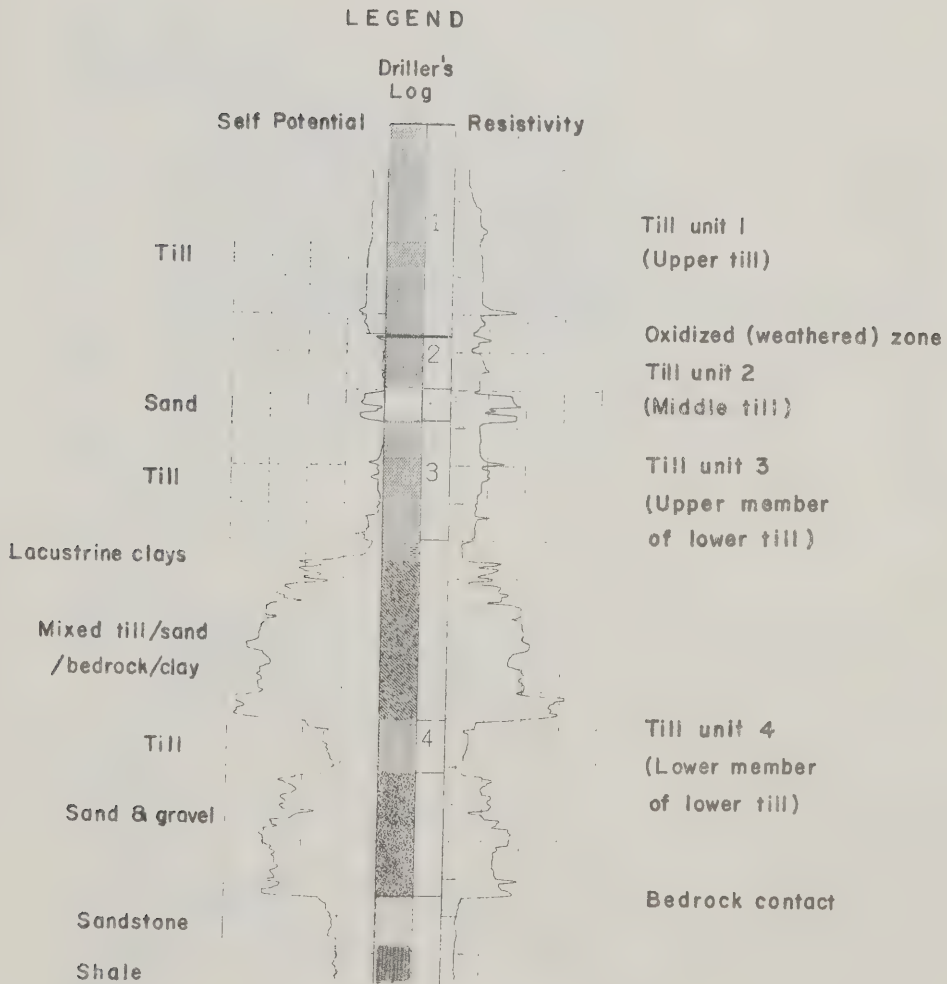
Hole/Depth	Grain Lithology of 1-2mm Fraction (%)					Bulk Chemistry of Silt and Clay Fraction (%)							
	Local	Acid	Basic	Carb.	Qtz.	Na	Mg	Al	Si	K	Ca	Ti	Fe
957E 30	23.5	54.5	3.5	2.5	16.0	0.42	1.41	7.73	32.49	2.10	3.03	0.43	3.90
957E 40	3.2	47.4	2.5	33.2	13.7	0.45	0.96	8.12	31.40	1.96	2.83	1.17	5.09
957E 50	15.5	59.5	4.0	1.5	19.5	0.53	0.57	7.16	35.15	1.87	1.79	0.32	3.14
957E 60	55.5	25.0	2.0	0.5	17.0	0.16	0.37	6.59	35.98	1.65	0.78	0.54	3.73
957E 70	36.0	32.5	2.0	11.5	18.0	0.0	0.42	7.30	35.72	1.73	0.94	0.46	3.41
957E 80	6.5	68.8	4.8	3.2	16.7	0.70	0.68	8.01	33.35	1.79	1.81	0.54	4.14
958E 10	11.0	62.0	3.0	4.0	20.0	0.17	0.99	8.11	32.85	2.02	2.50	0.48	4.12
958E 20	51.7	30.2	2.6	0.9	14.7	0.0	0.85	7.95	33.66	1.93	1.90	0.45	4.23
958E 30	6.0	58.7	5.3	2.7	27.3	0.19	0.76	8.20	33.72	1.94	1.69	0.50	3.76
958E 40	43.5	34.5	0.5	0.5	21.0	0.27	0.84	7.65	33.99	1.87	1.84	0.48	3.82
958E 50	3.7	62.0	2.5	5.5	26.4	0.42	0.94	7.90	33.01	2.53	2.69	0.39	3.53
958E 57	7.7	56.4	1.9	0.6	33.3	0.52	1.09	7.78	32.96	2.08	2.38	0.56	3.85



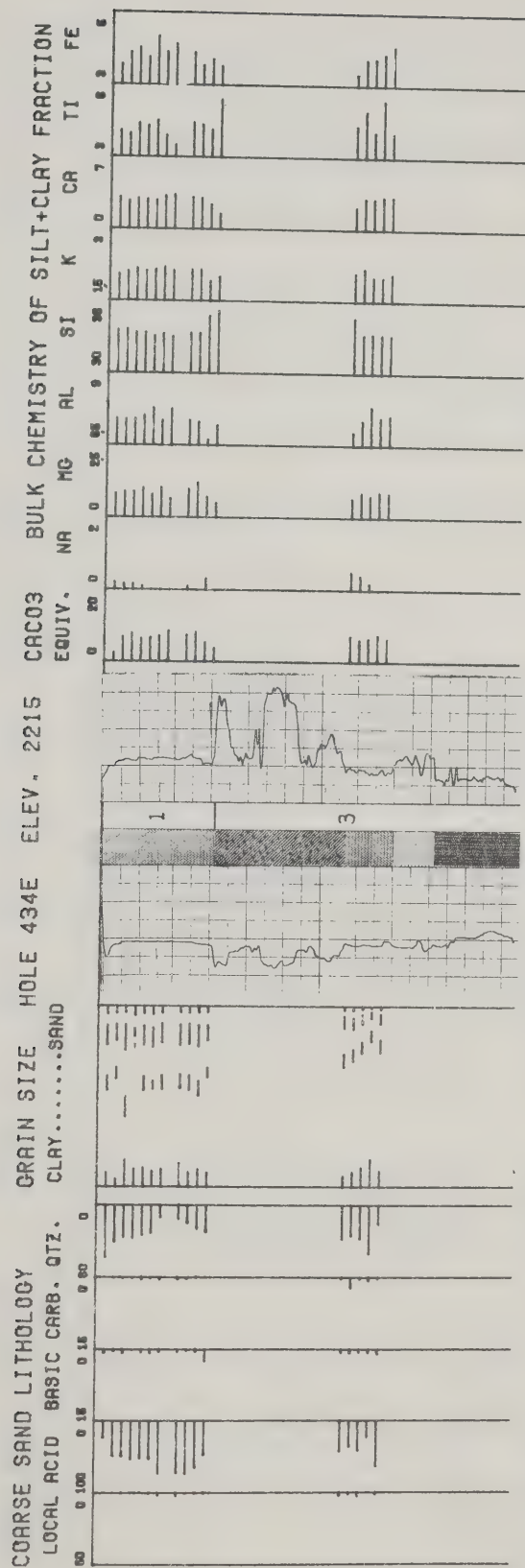


## APPENDIX 4

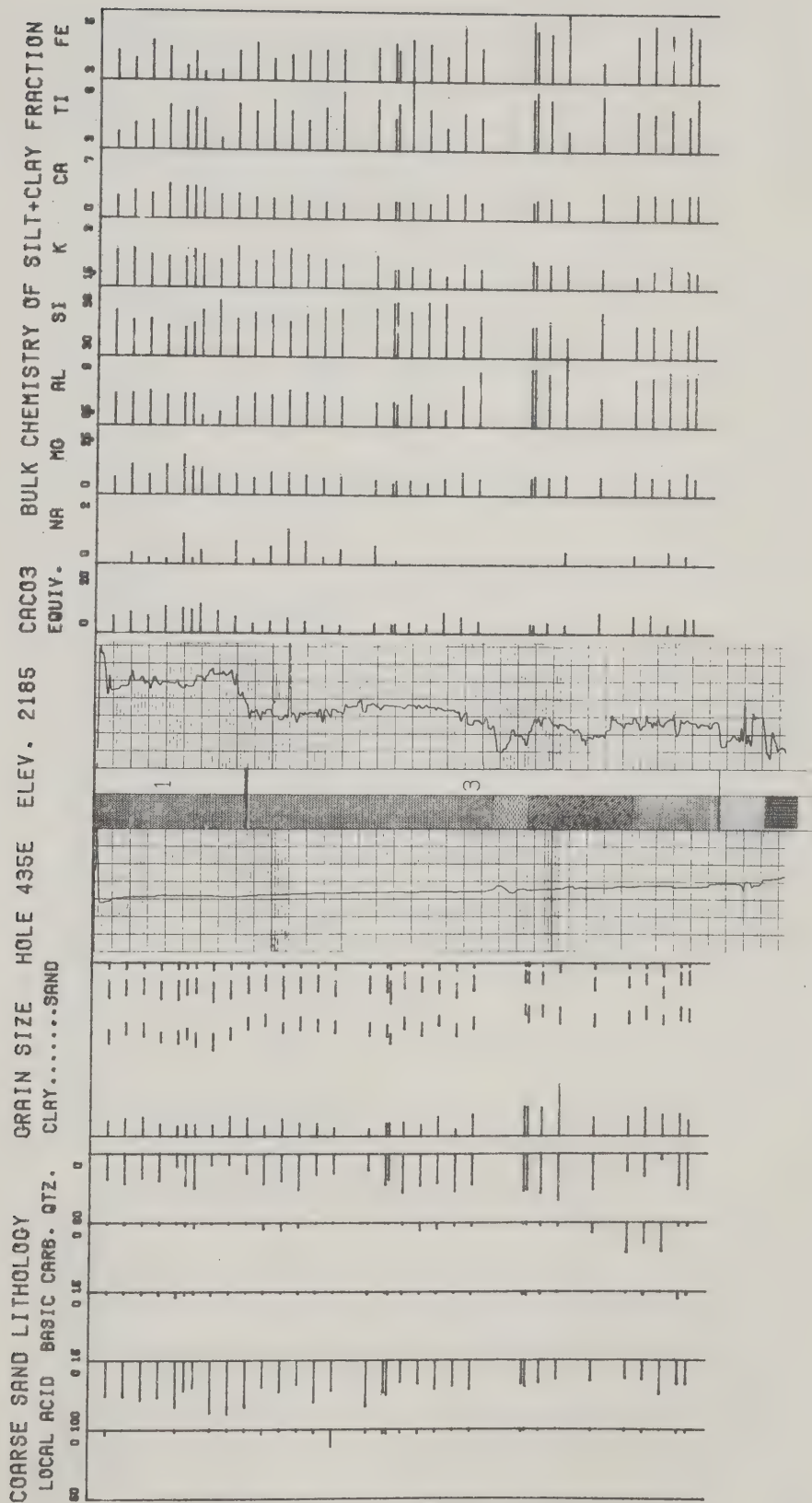
Plots of each drill hole, showing all analyzed variables, lithology and stratigraphic interpretation.



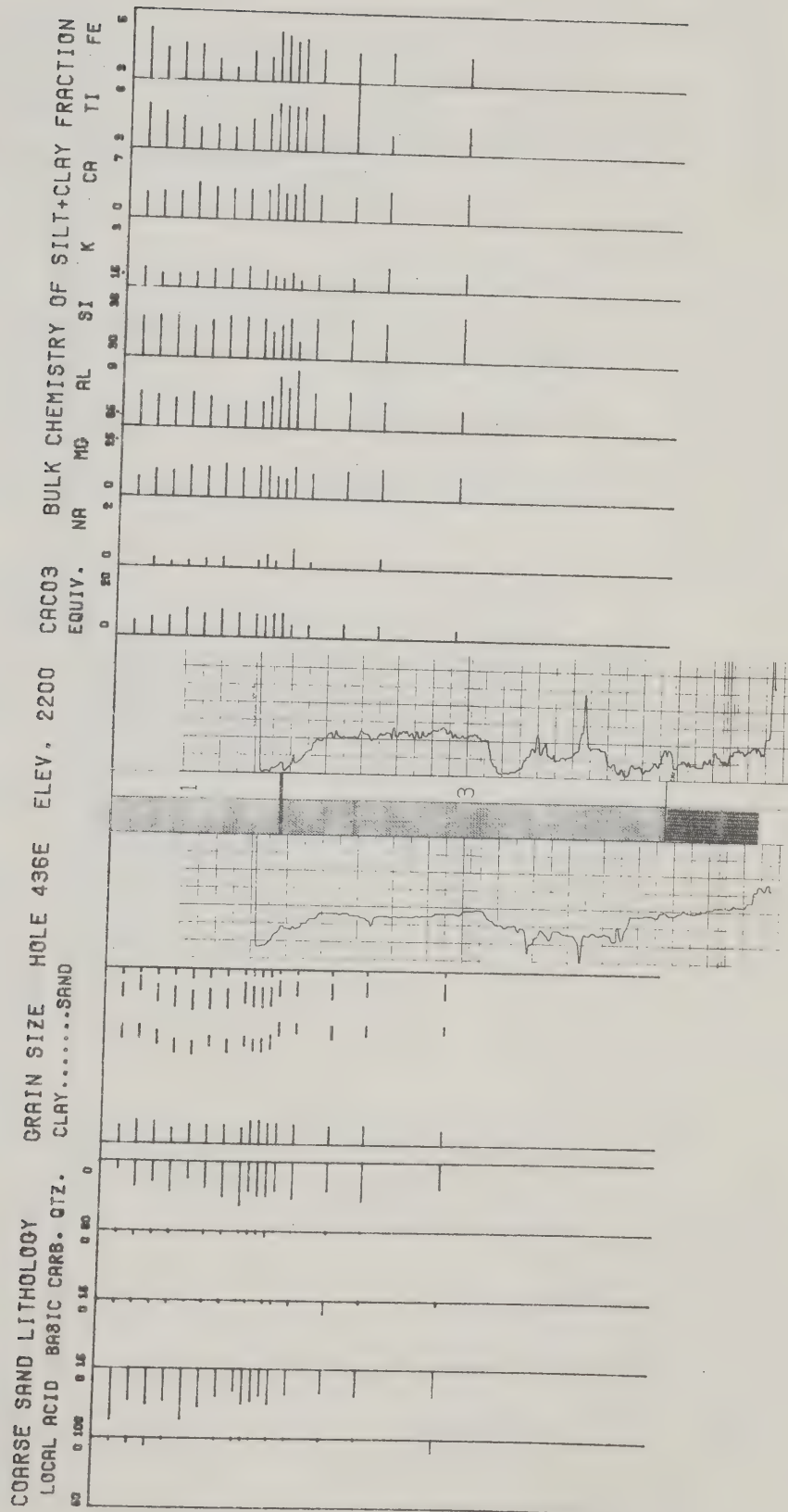






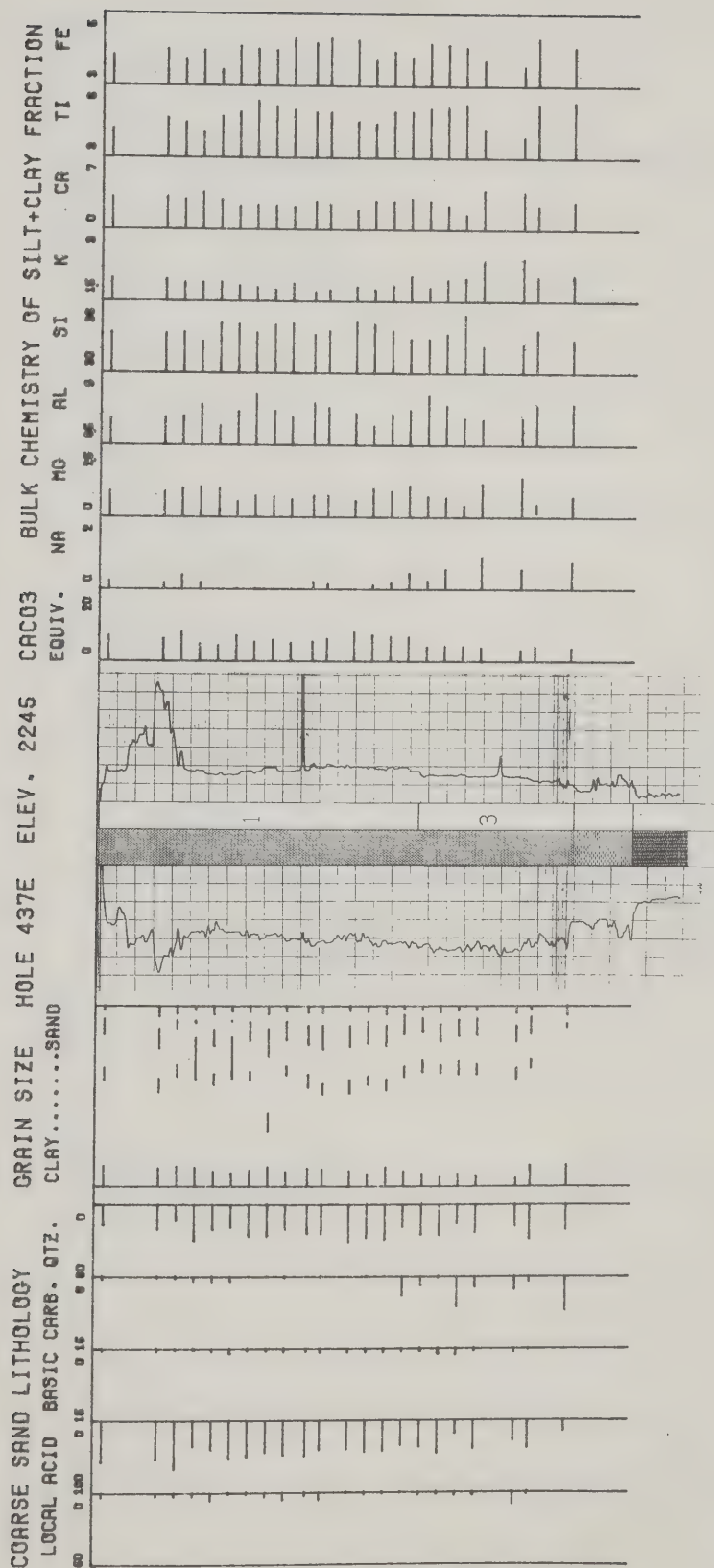




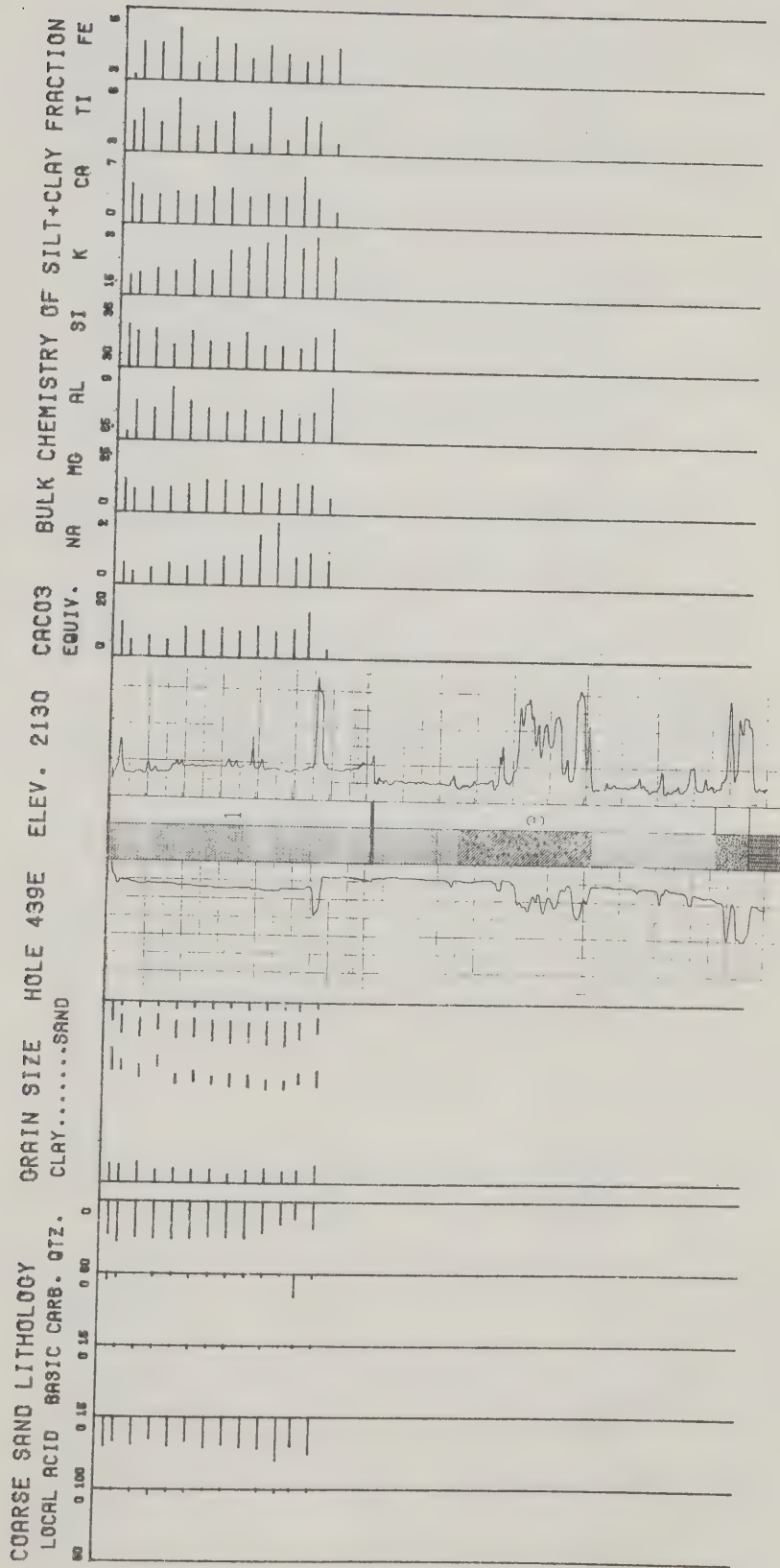




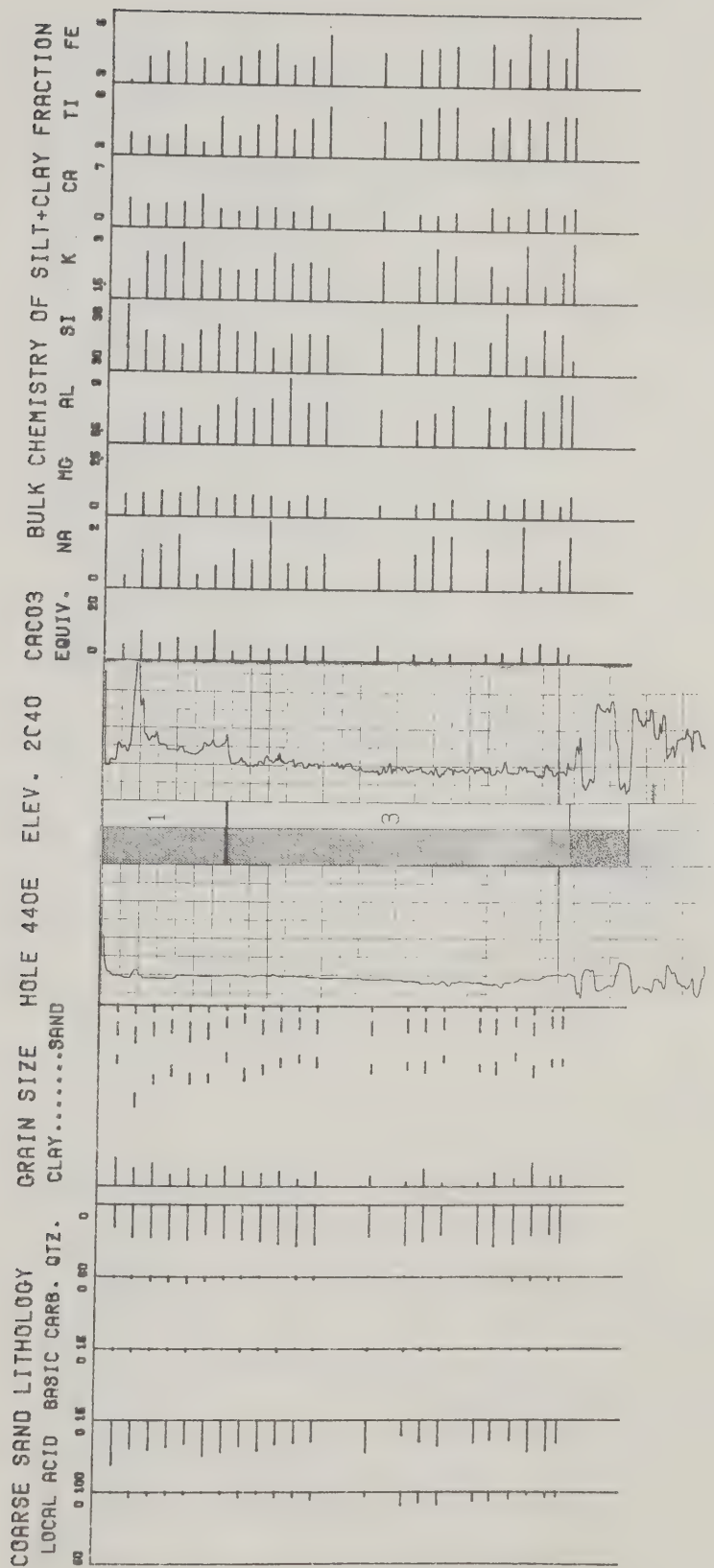




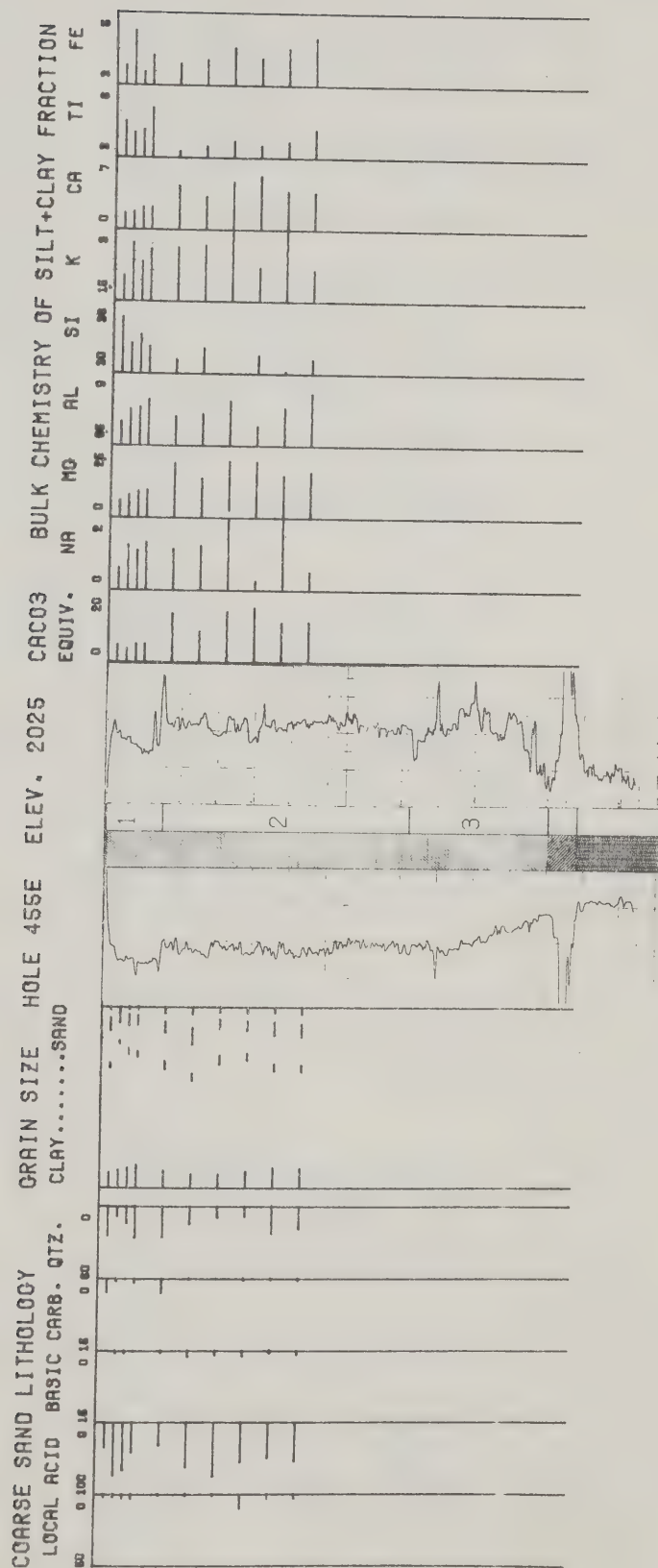






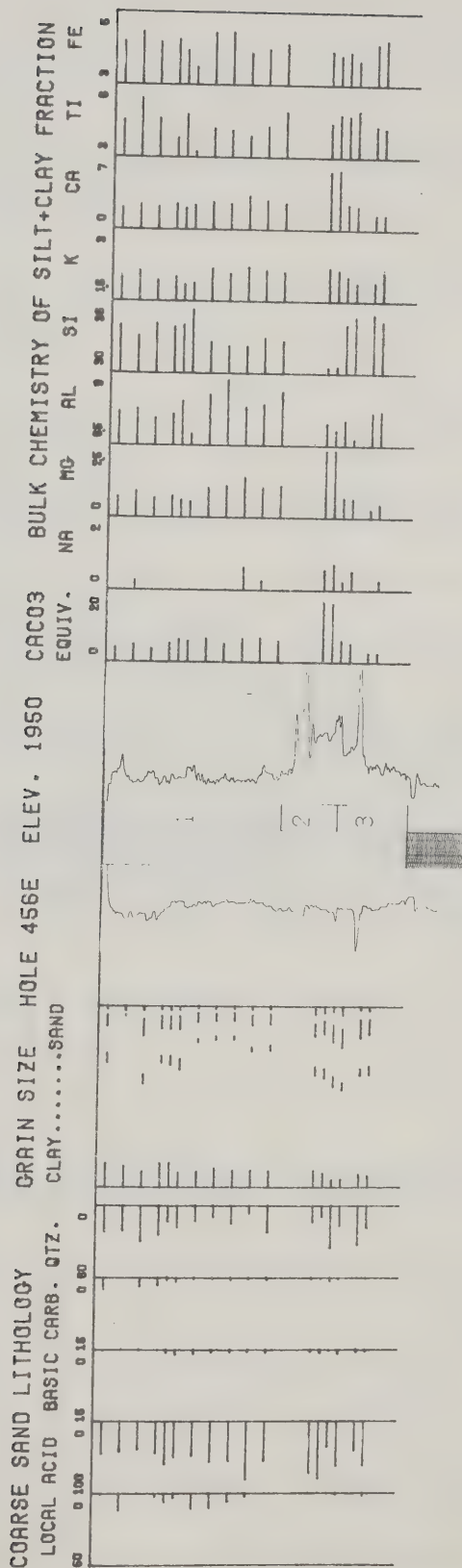




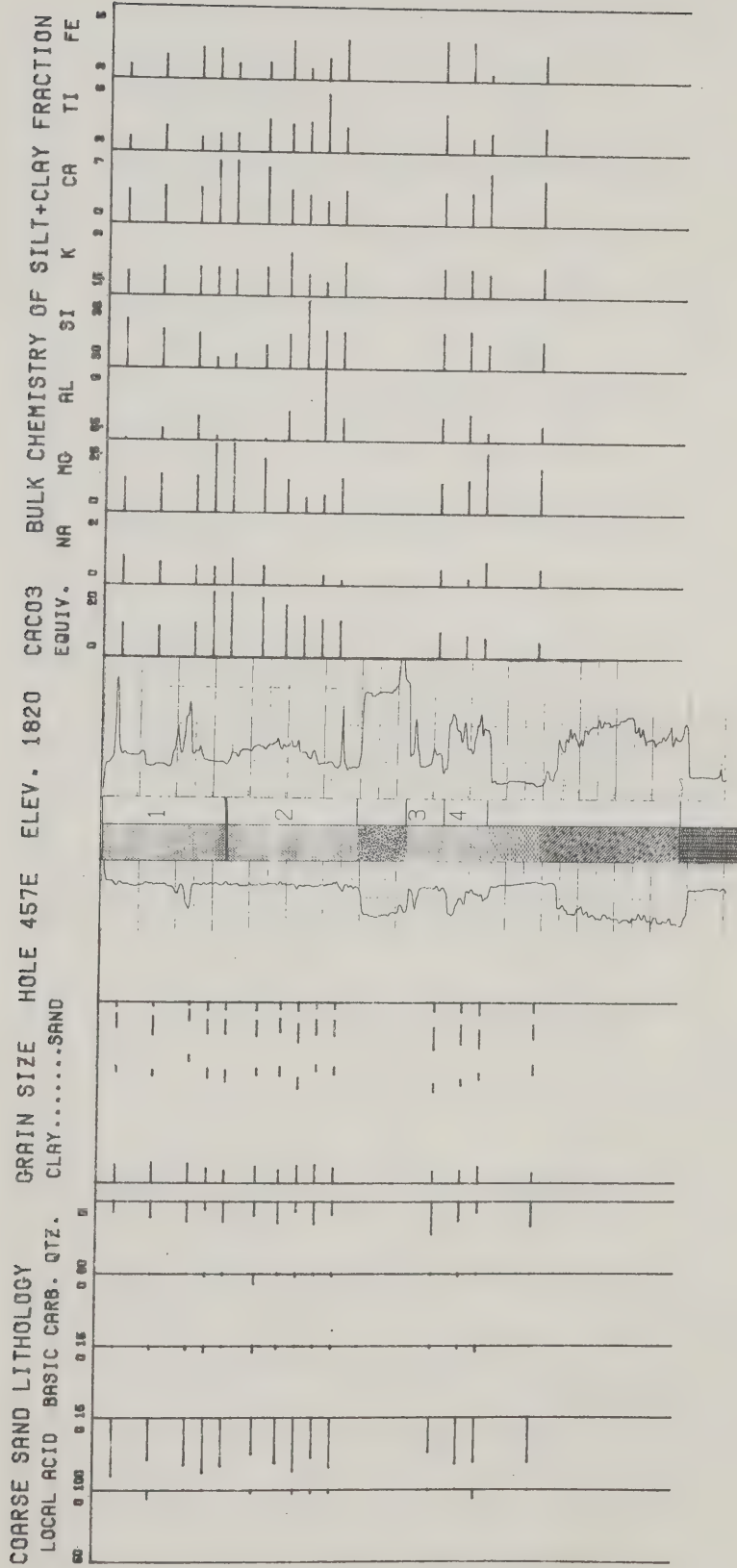




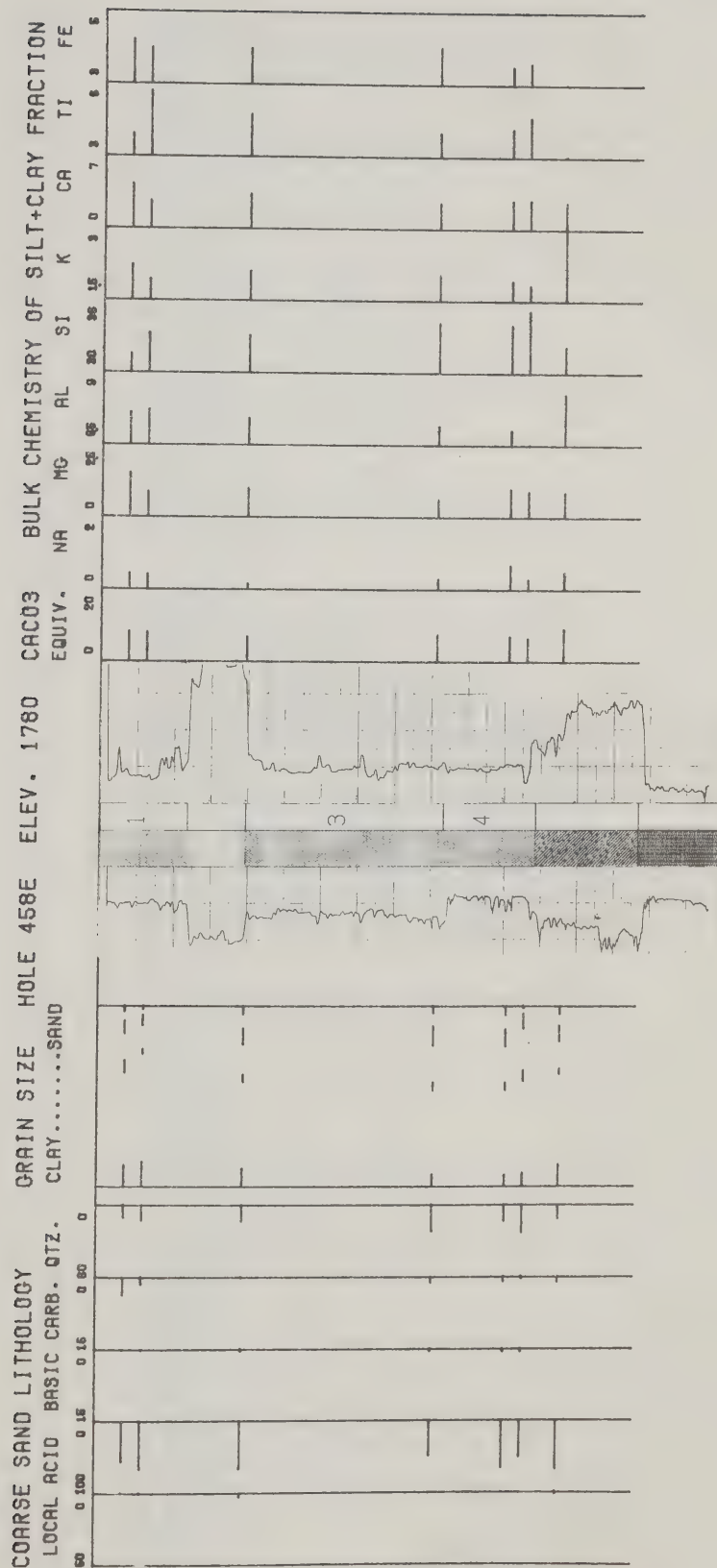




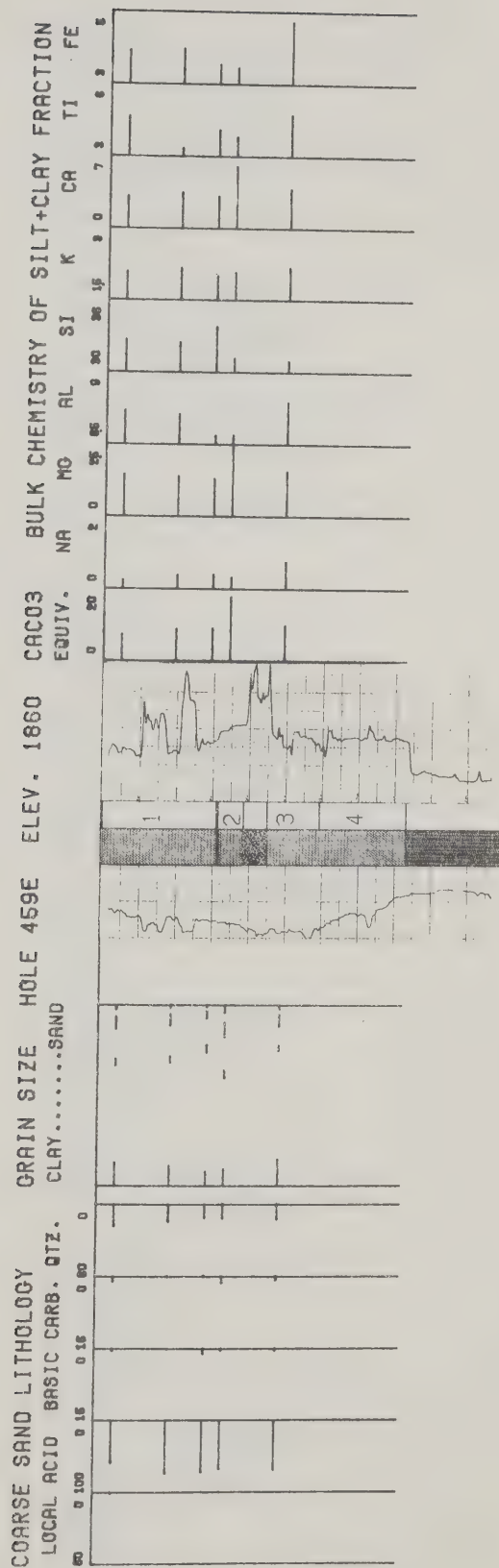






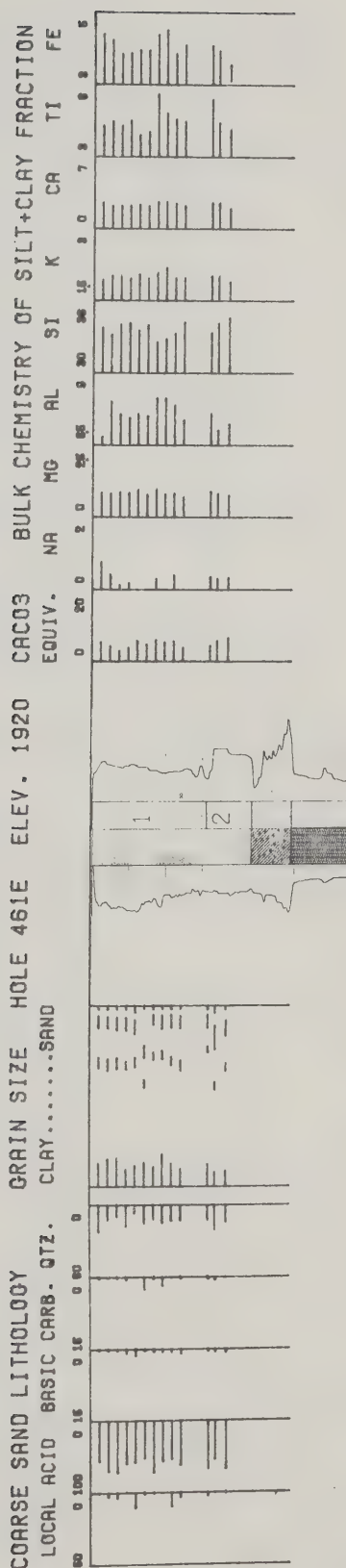




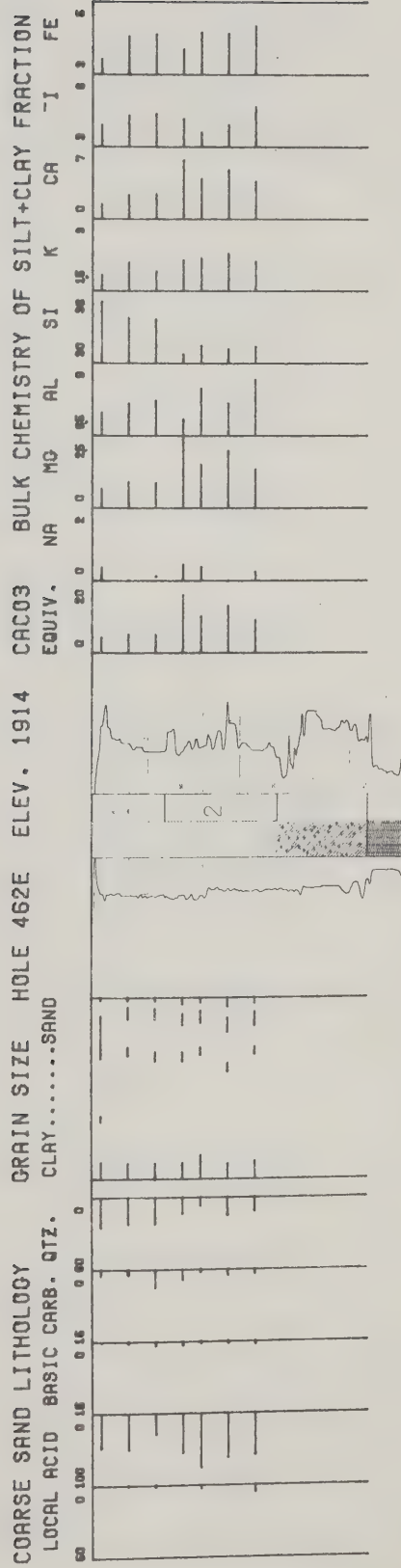




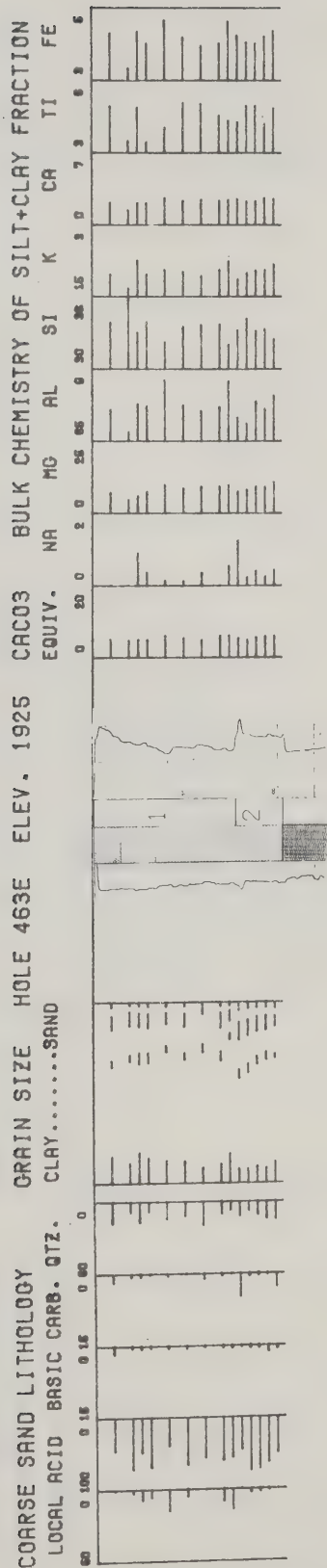




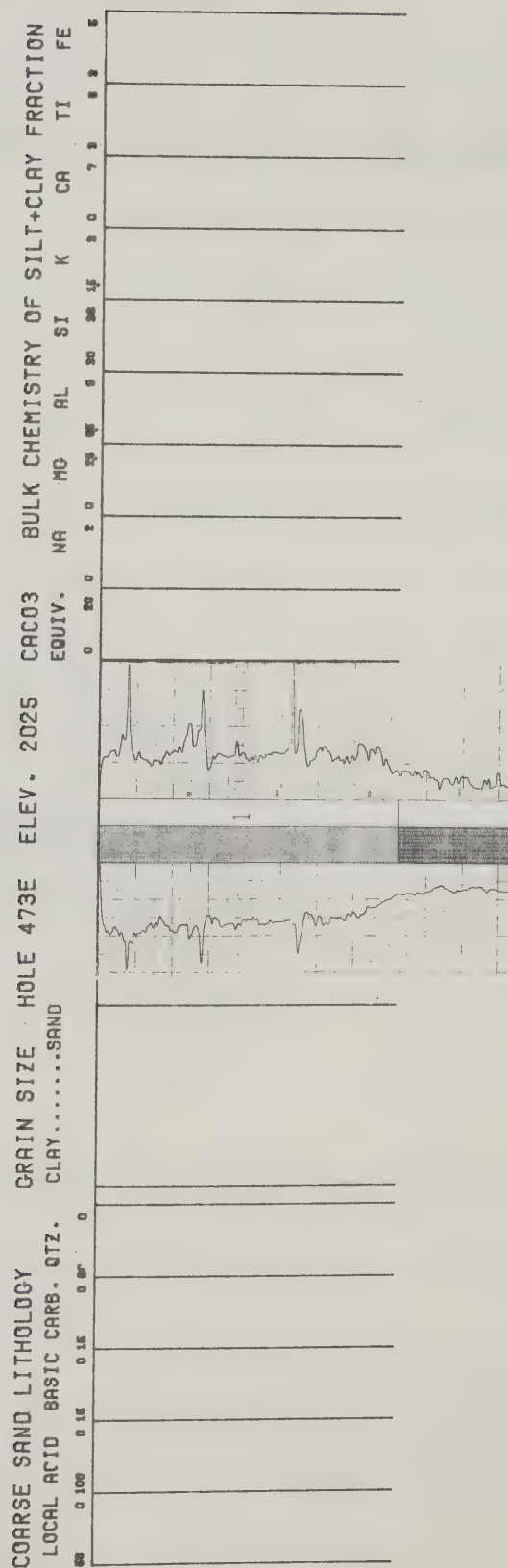






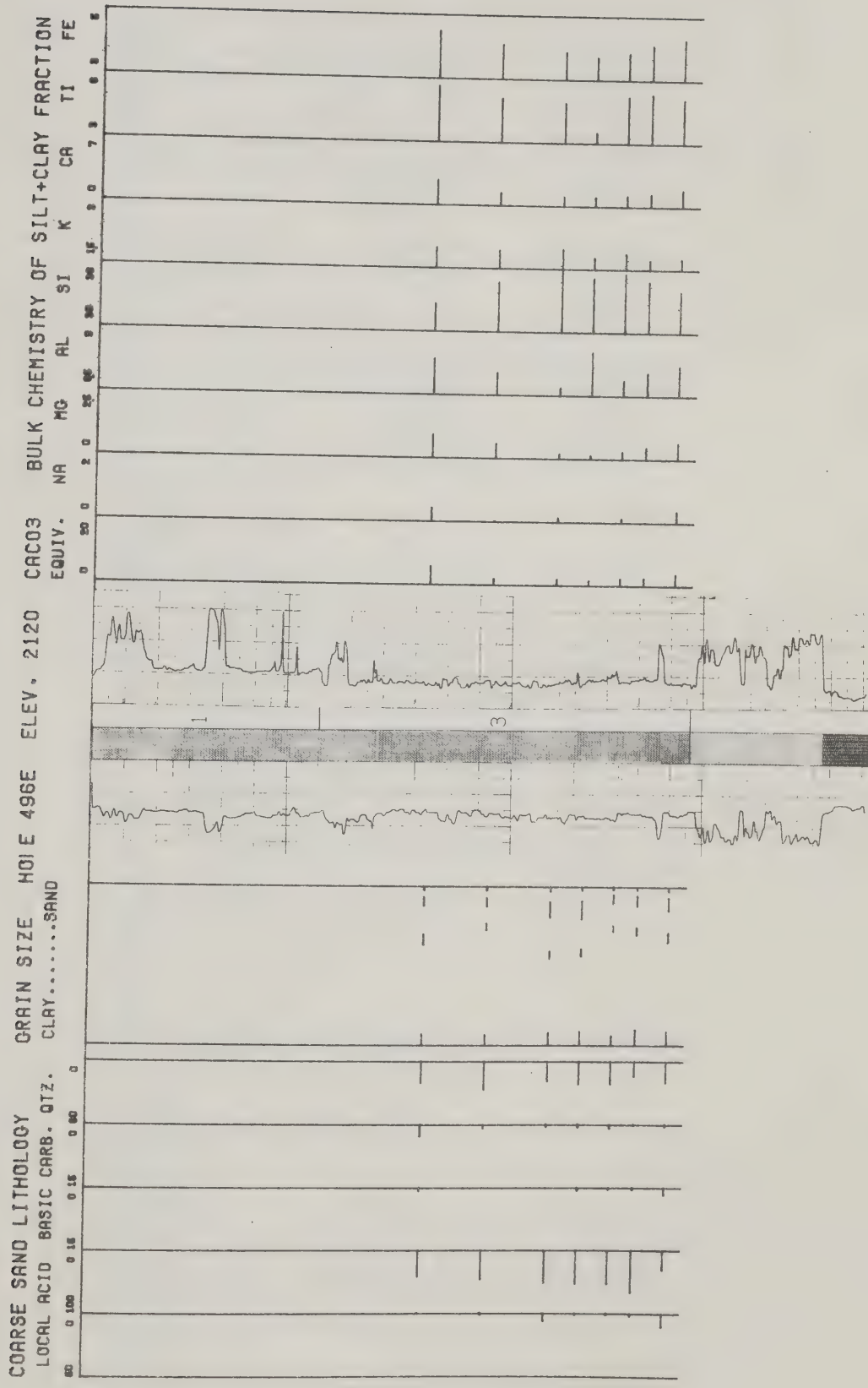




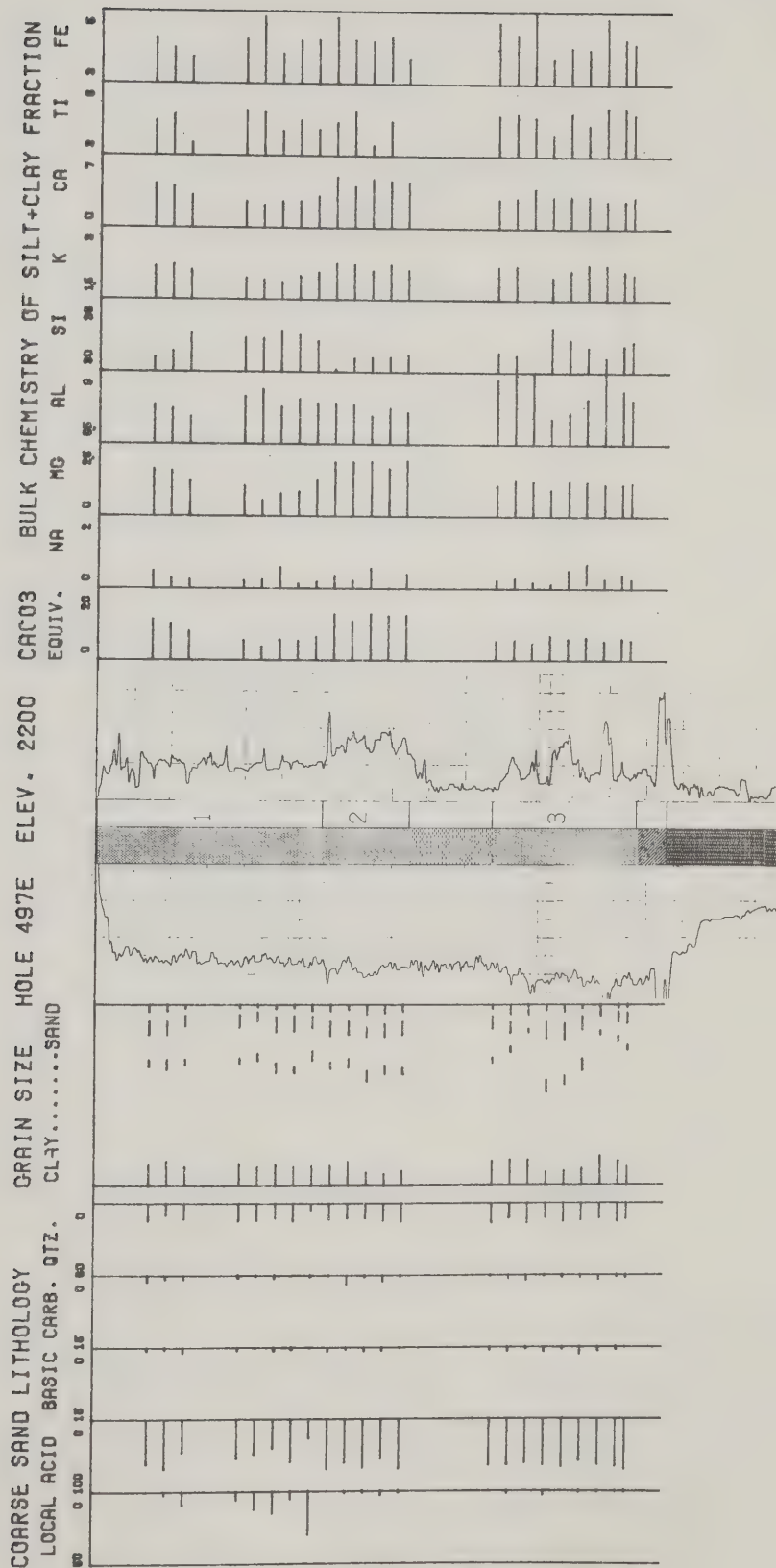




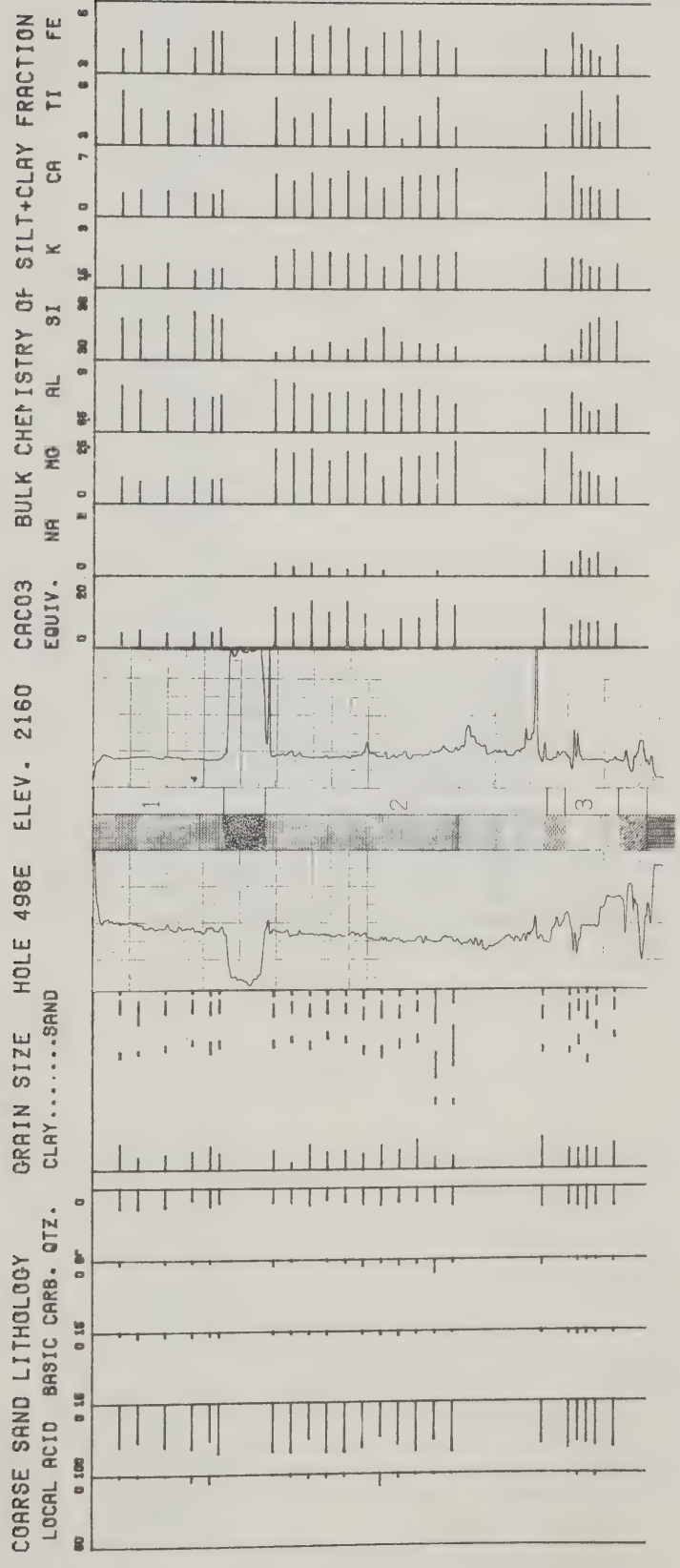






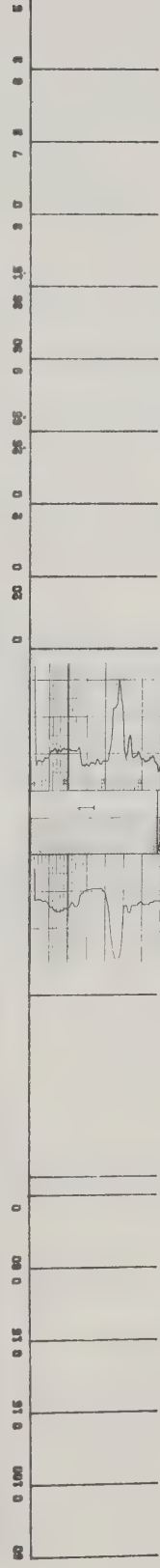






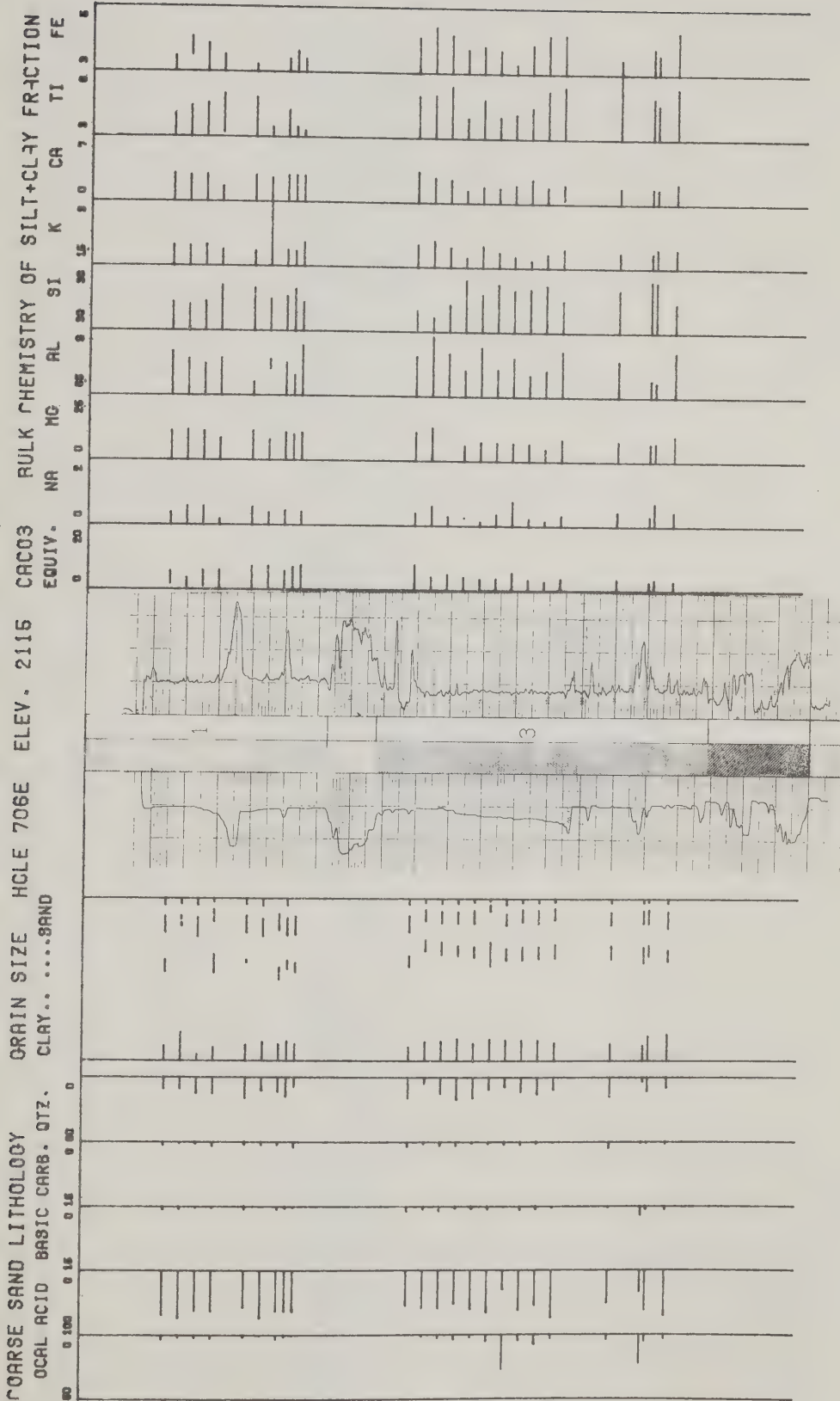


COARSE SAND LITHOLOGY GRAIN SIZE HOLE 704E ELEV. 2065 BULK CHEMISTRY OF SILT+CLAY FRACTION





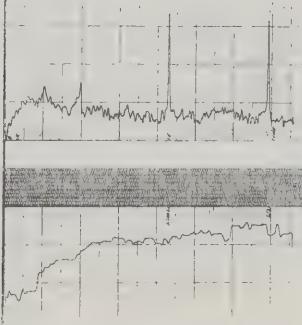




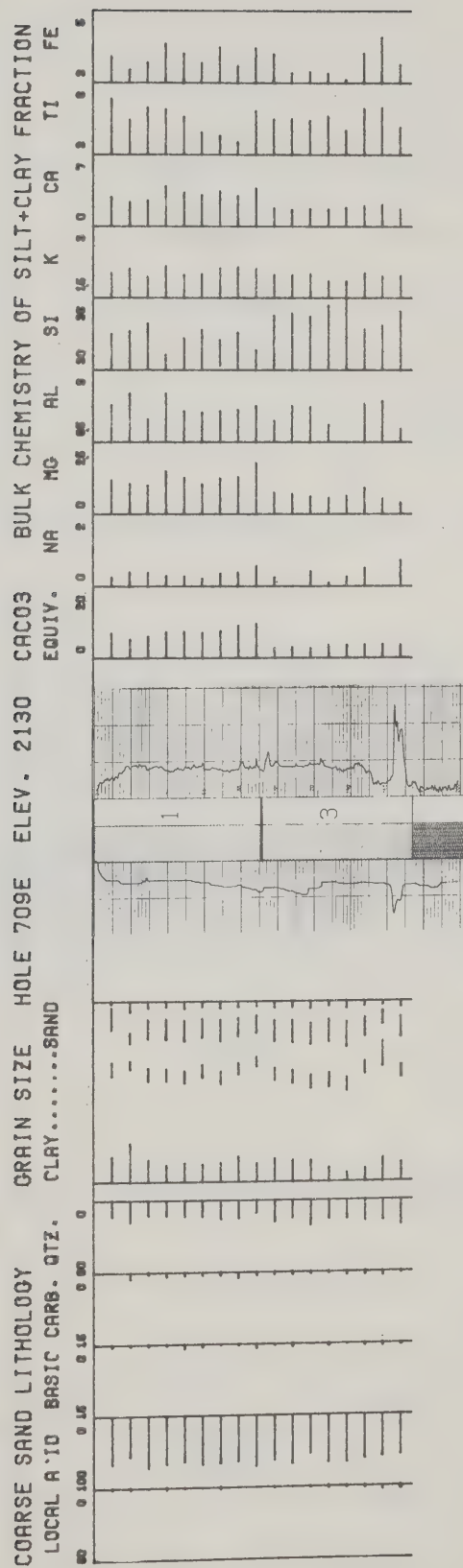


COARSE SAND LITHOLOGY GRAIN SIZE HOLE 707E ELEV. 1960 BULK CHEMISTRY OF SILT+CLAY FRACTION  
 LOCAL ACID BASIC CARB. QTZ. CLAY.....SAND EQUIV. NA MG -L SI K CA TI FE

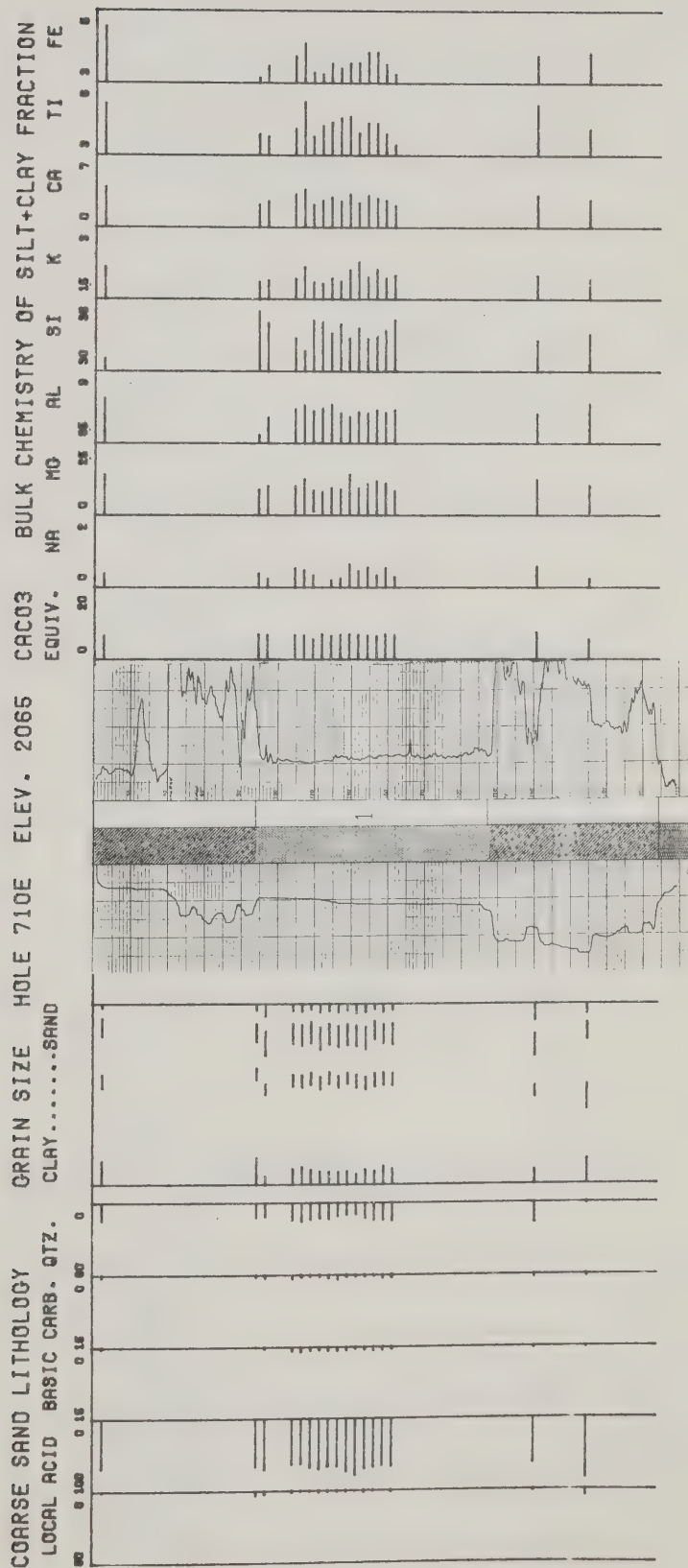
0.100 0.16 0.18 0.25 0 0 20.0 2.0 25 45 9 90 200 15 3.0 7.3 6.3 5





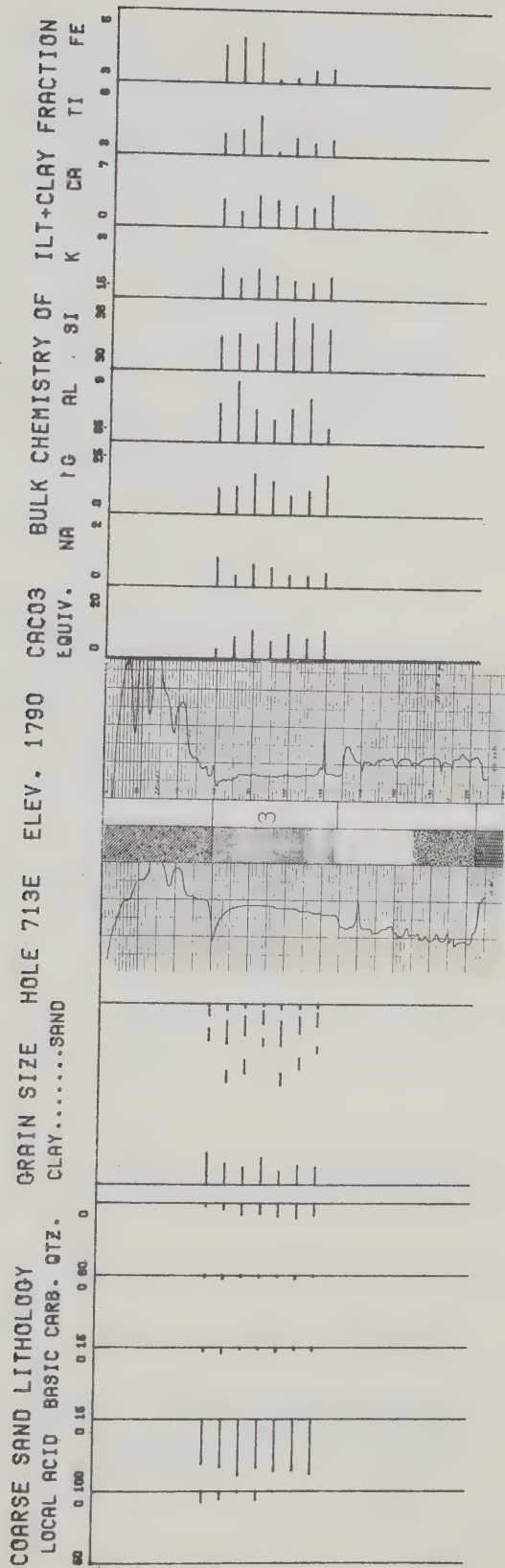




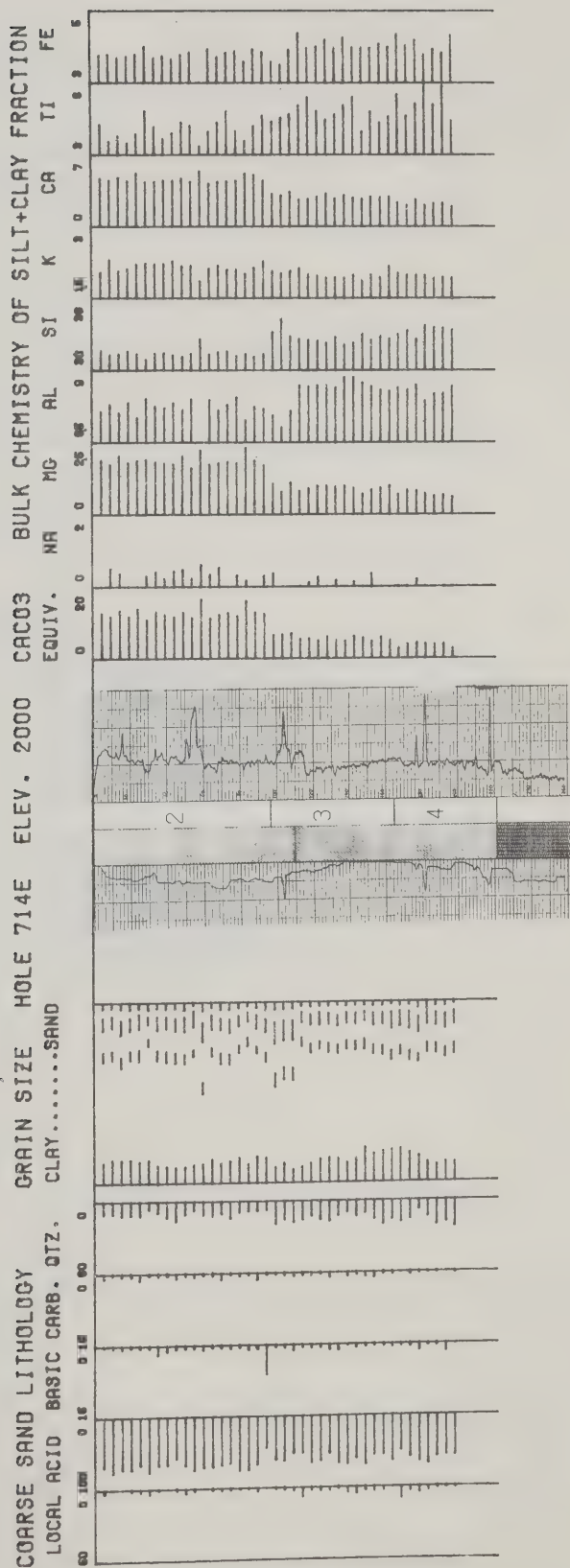




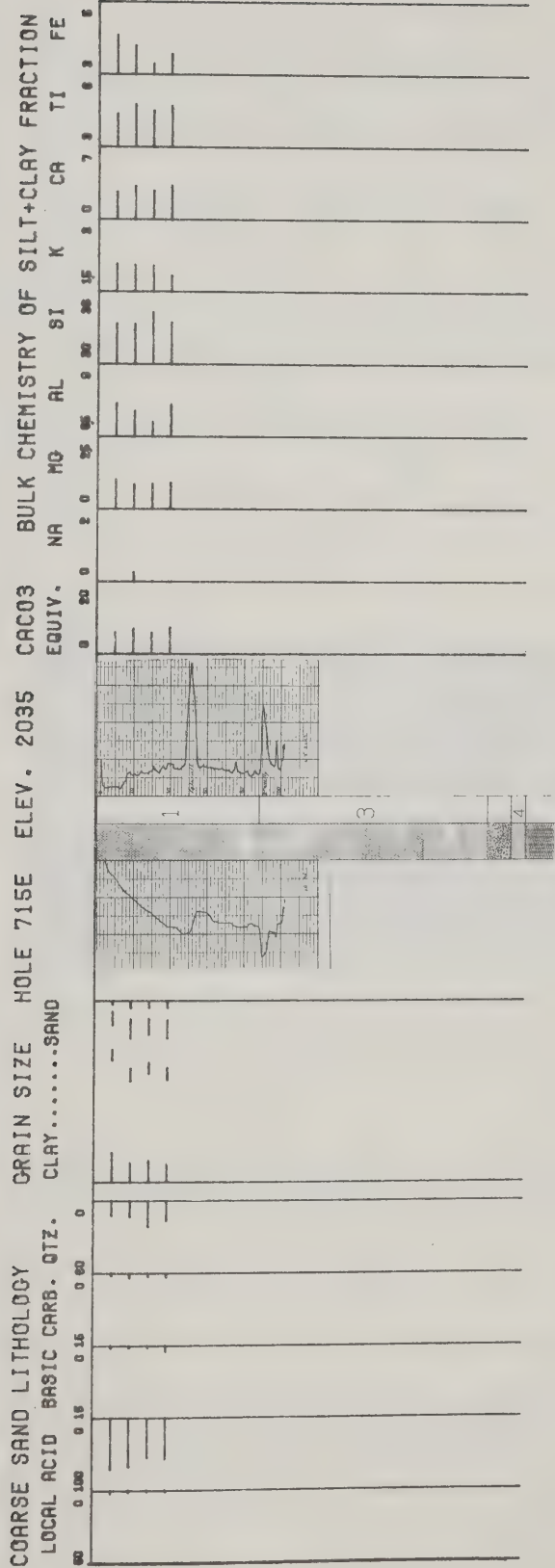




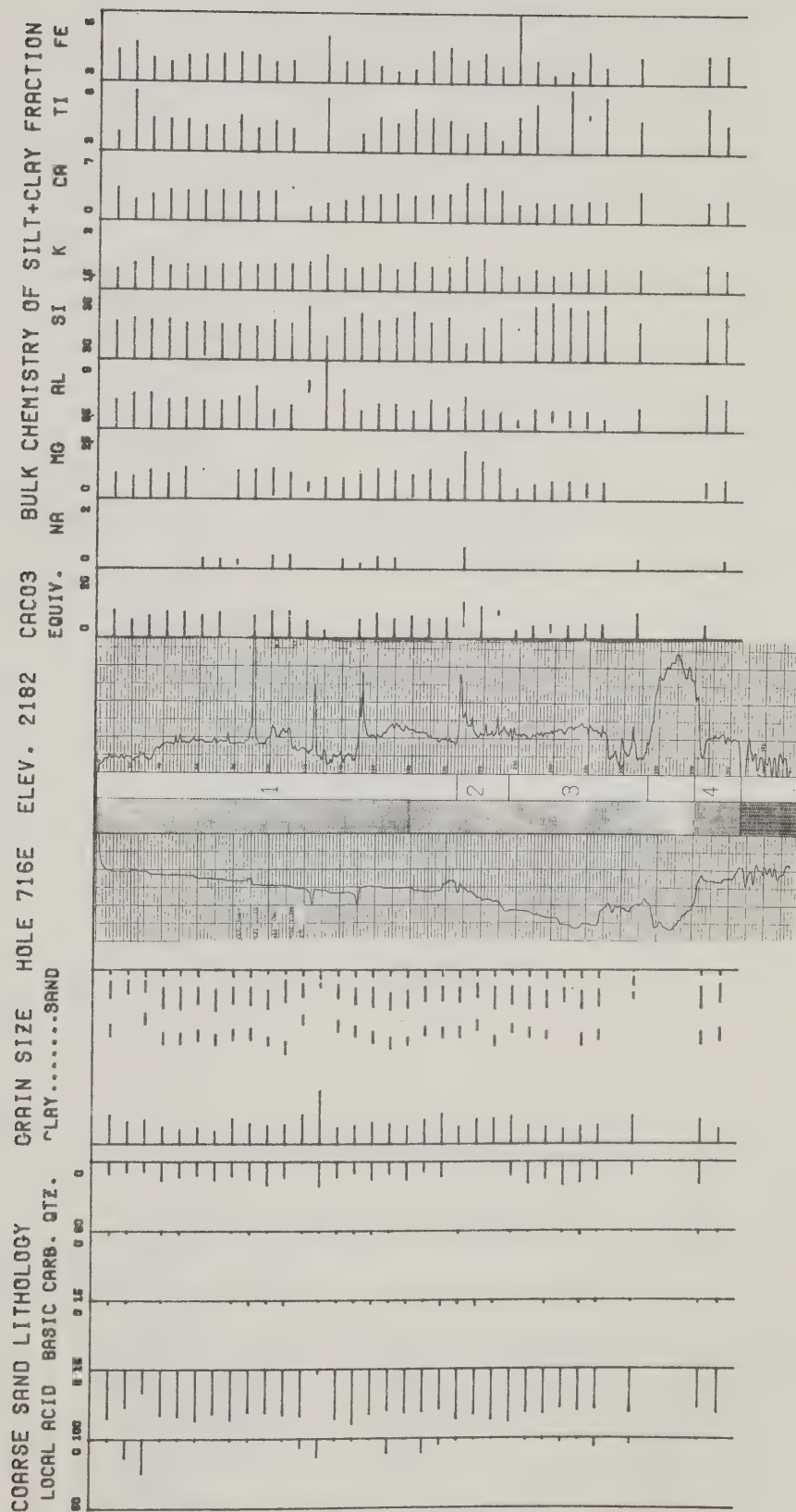






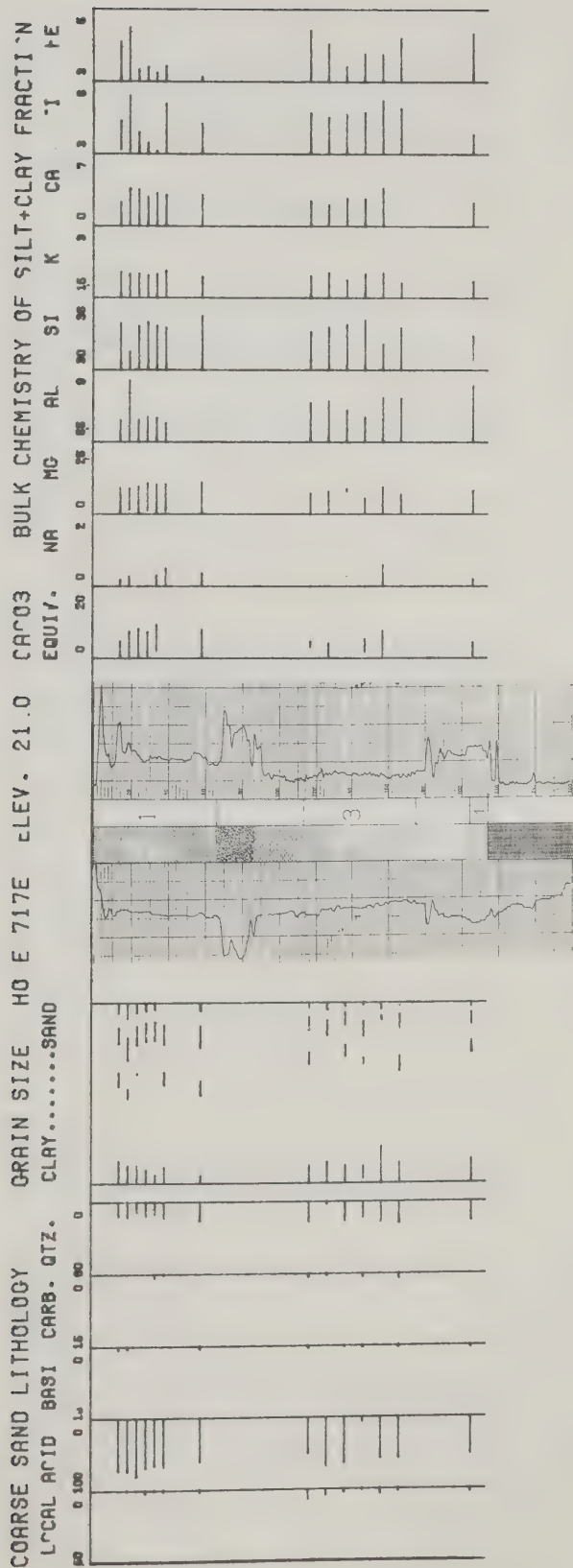




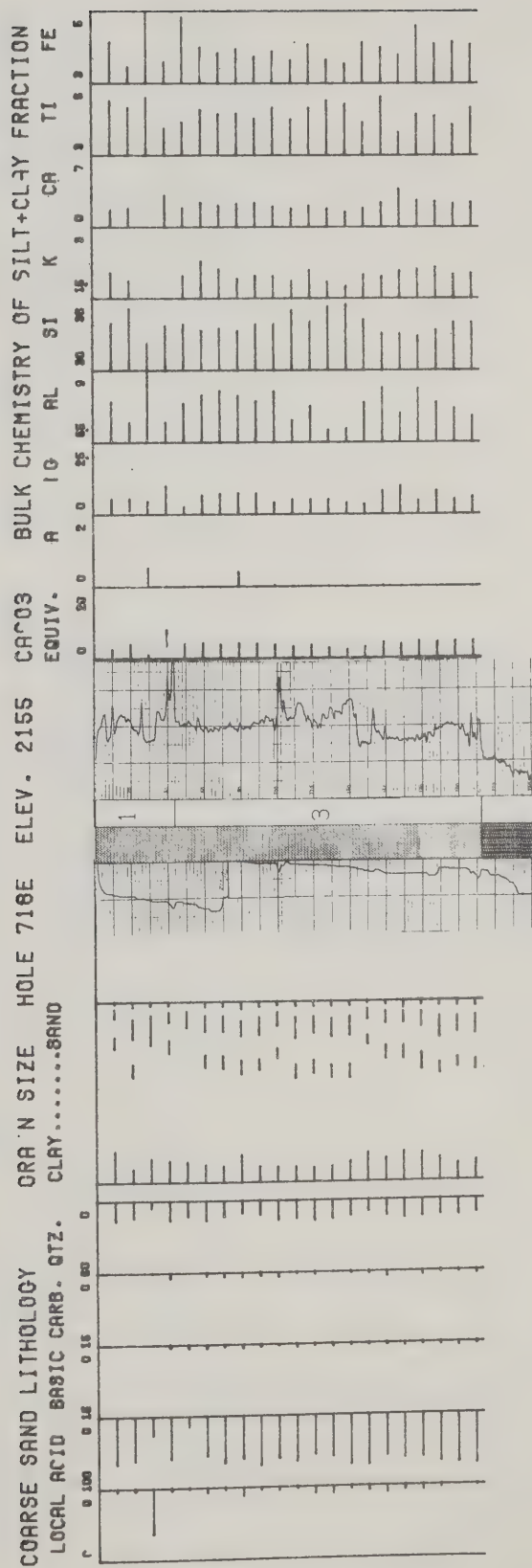




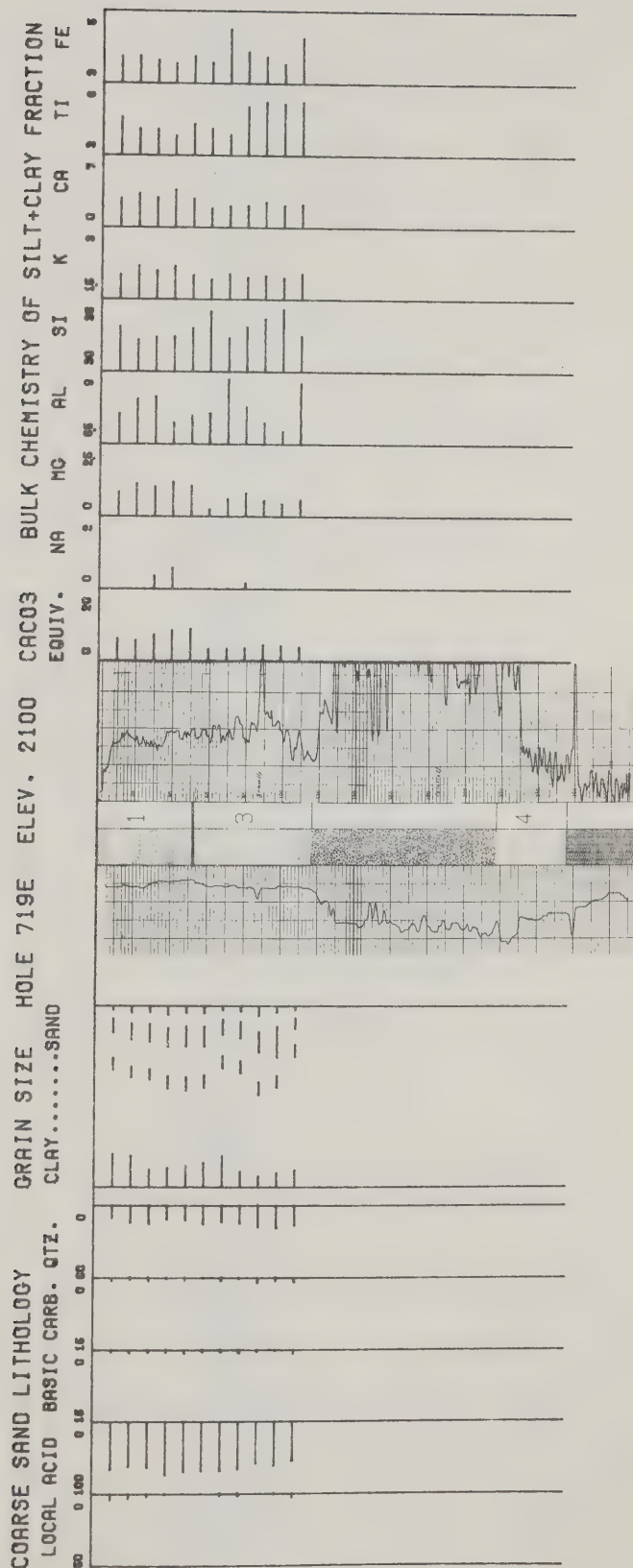












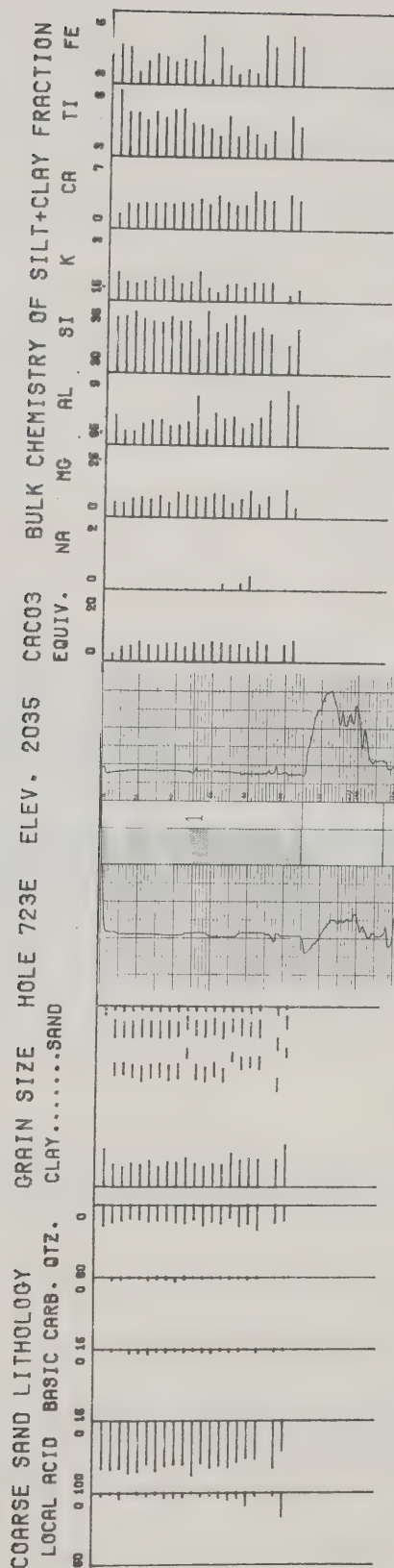


COARSE SAND LITHOLOGY GRAIN SIZE HOLE 721E ELEV. 1900 CAC03 BULK CHEMISTRY OF SILT+CLAY FRACTION  
LOCAL ACID BASIC CARB. QTZ. CLAY.....SAND EQUIV. NA MG AL SI K CA TI FE

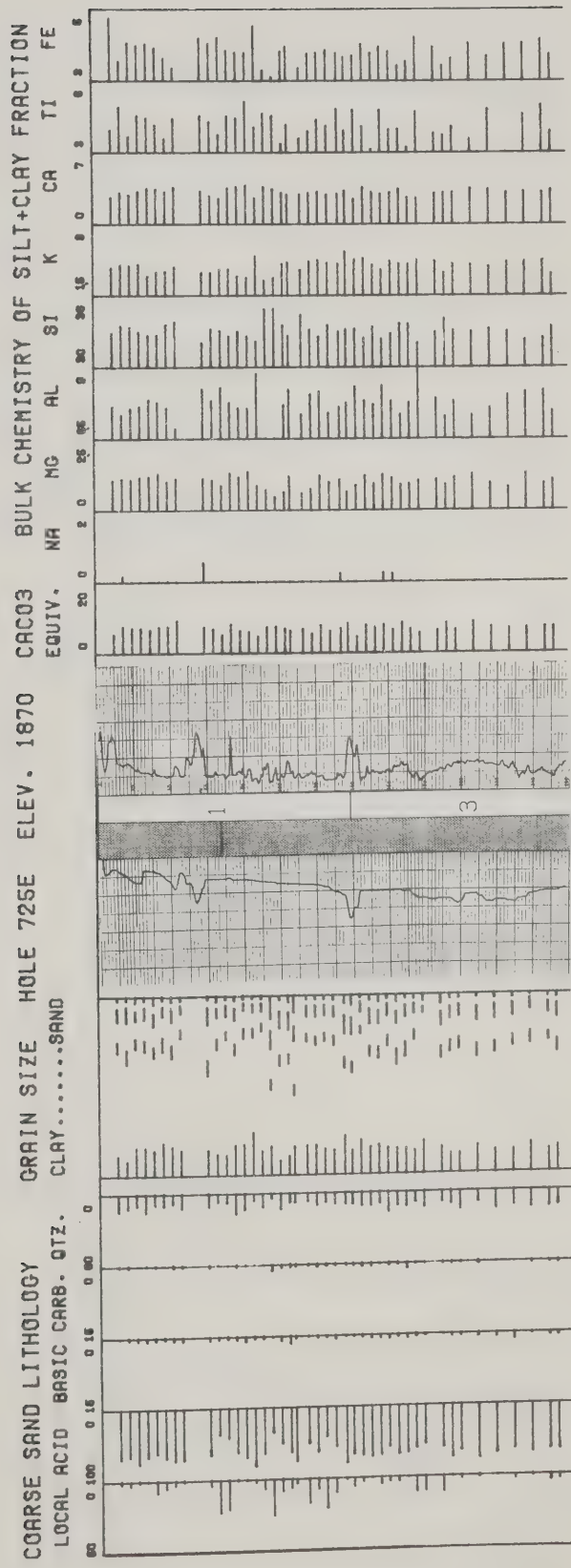




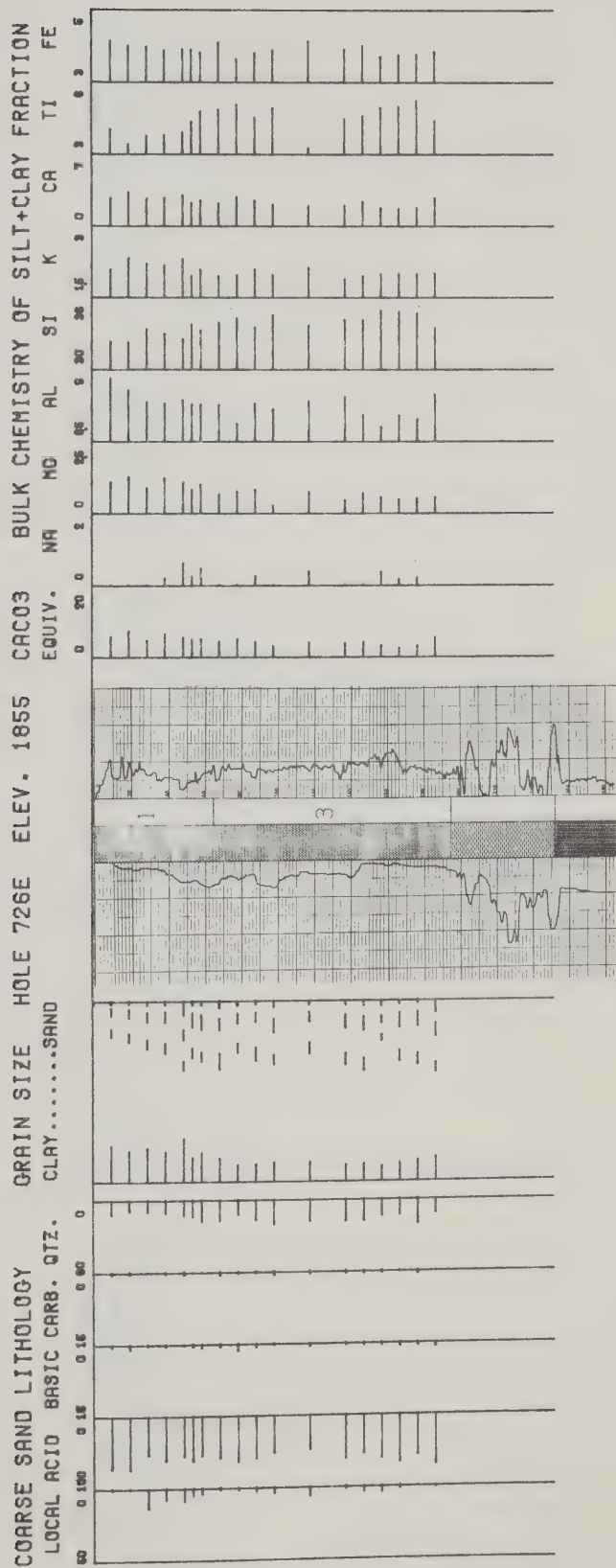




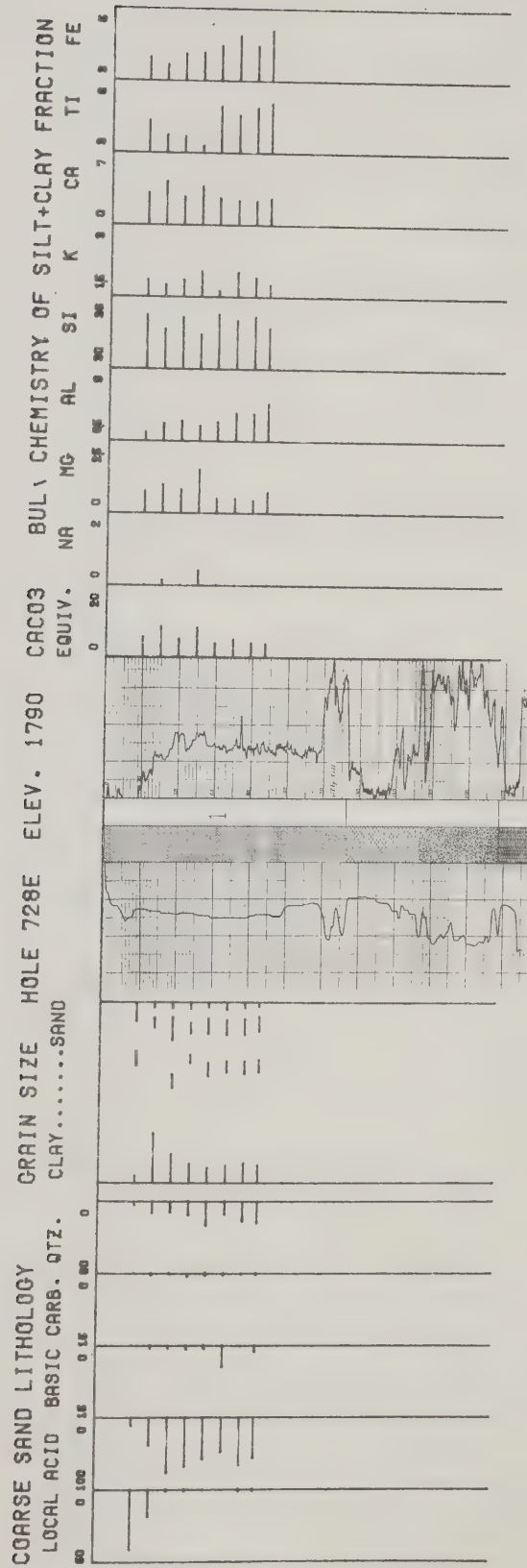






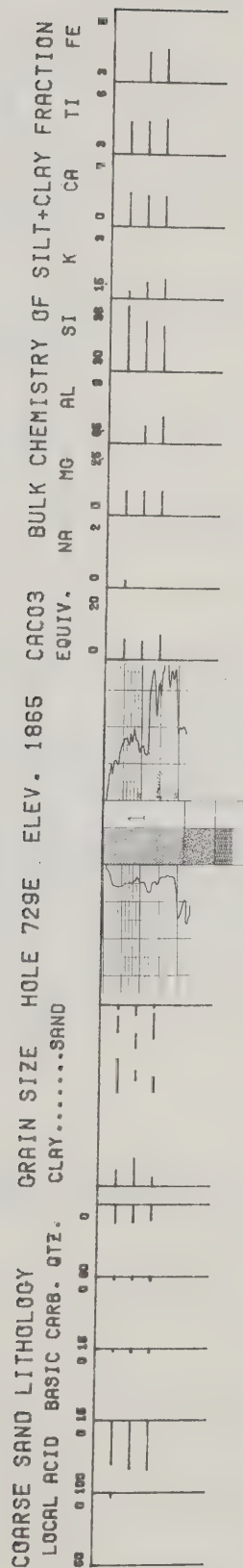




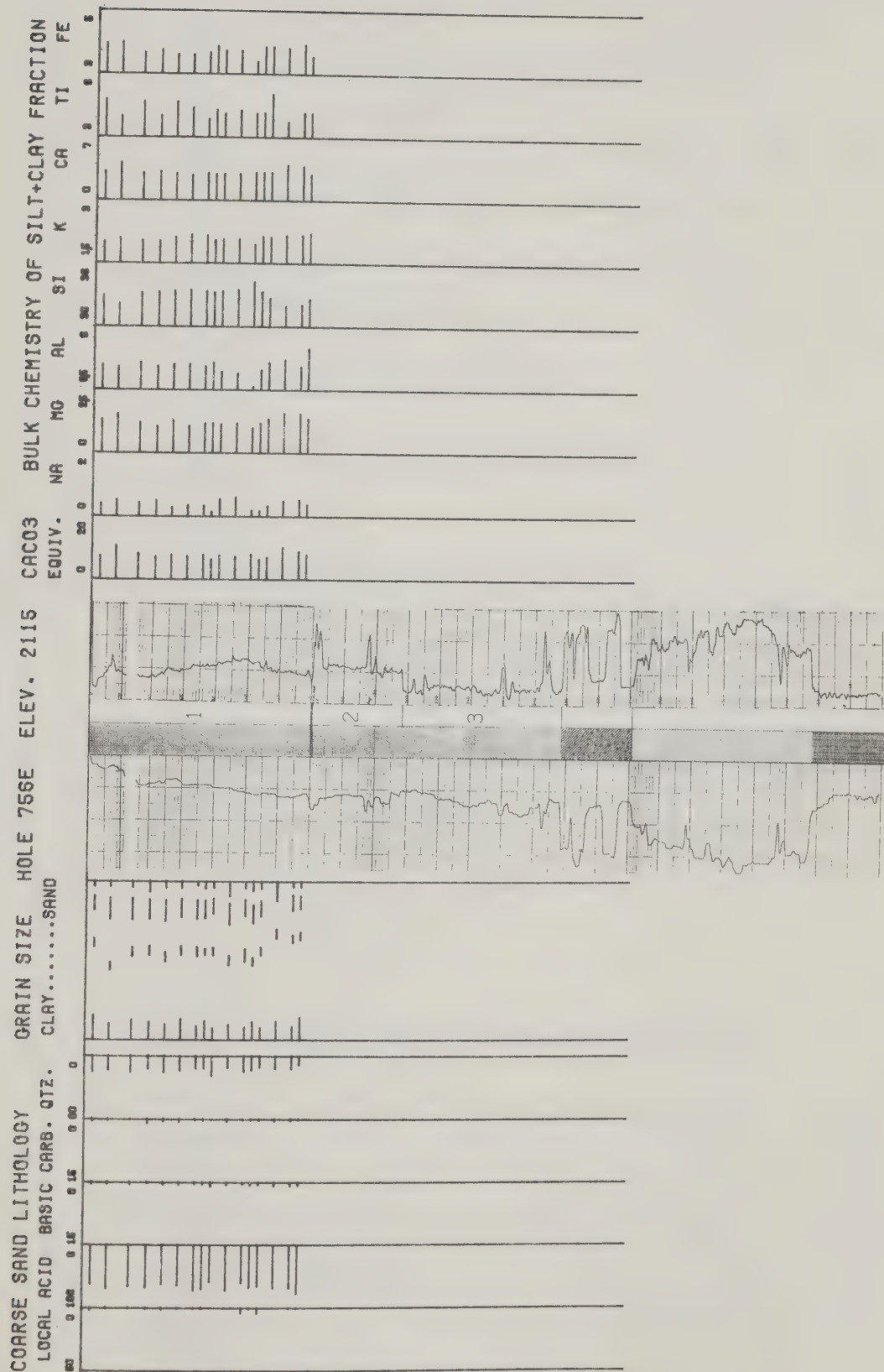




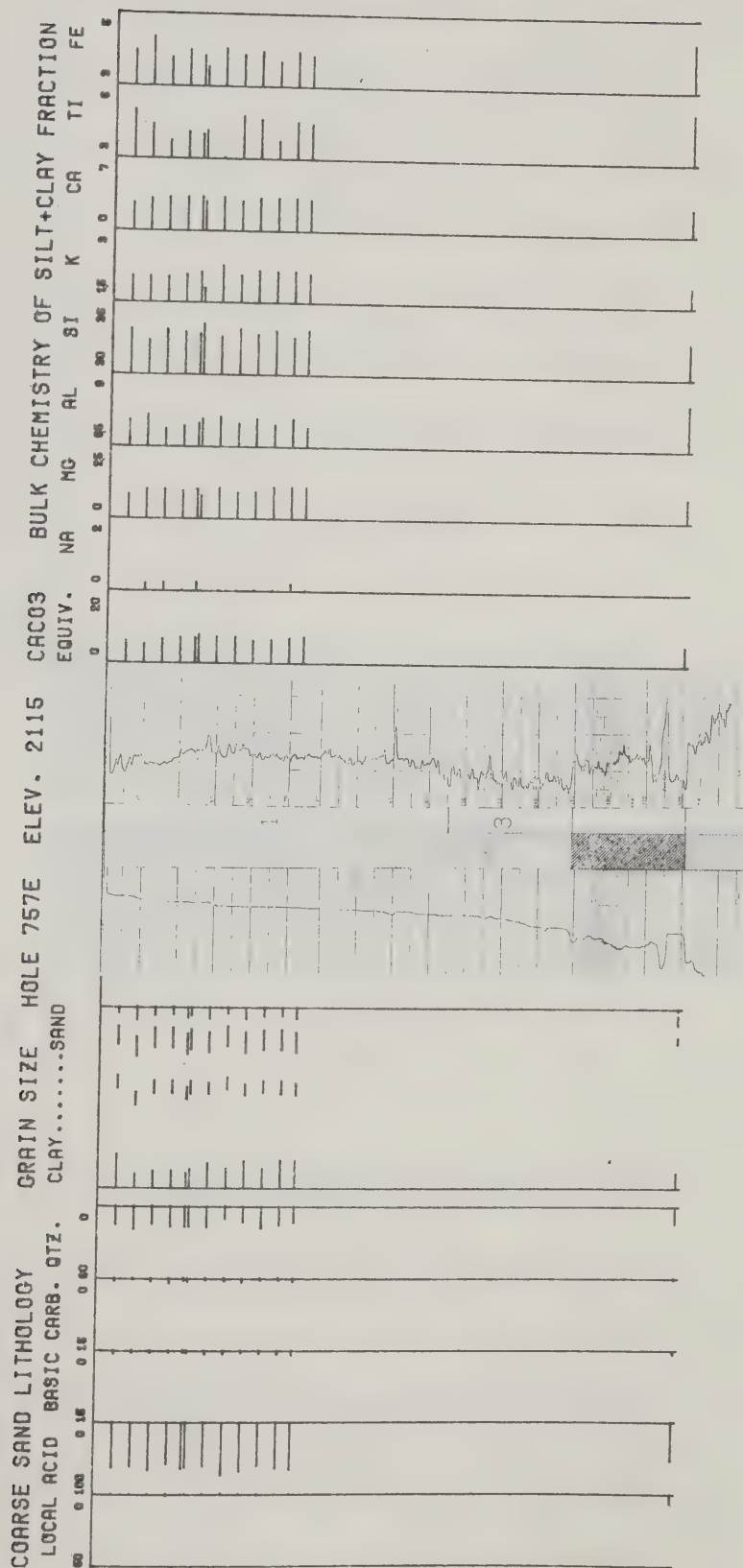




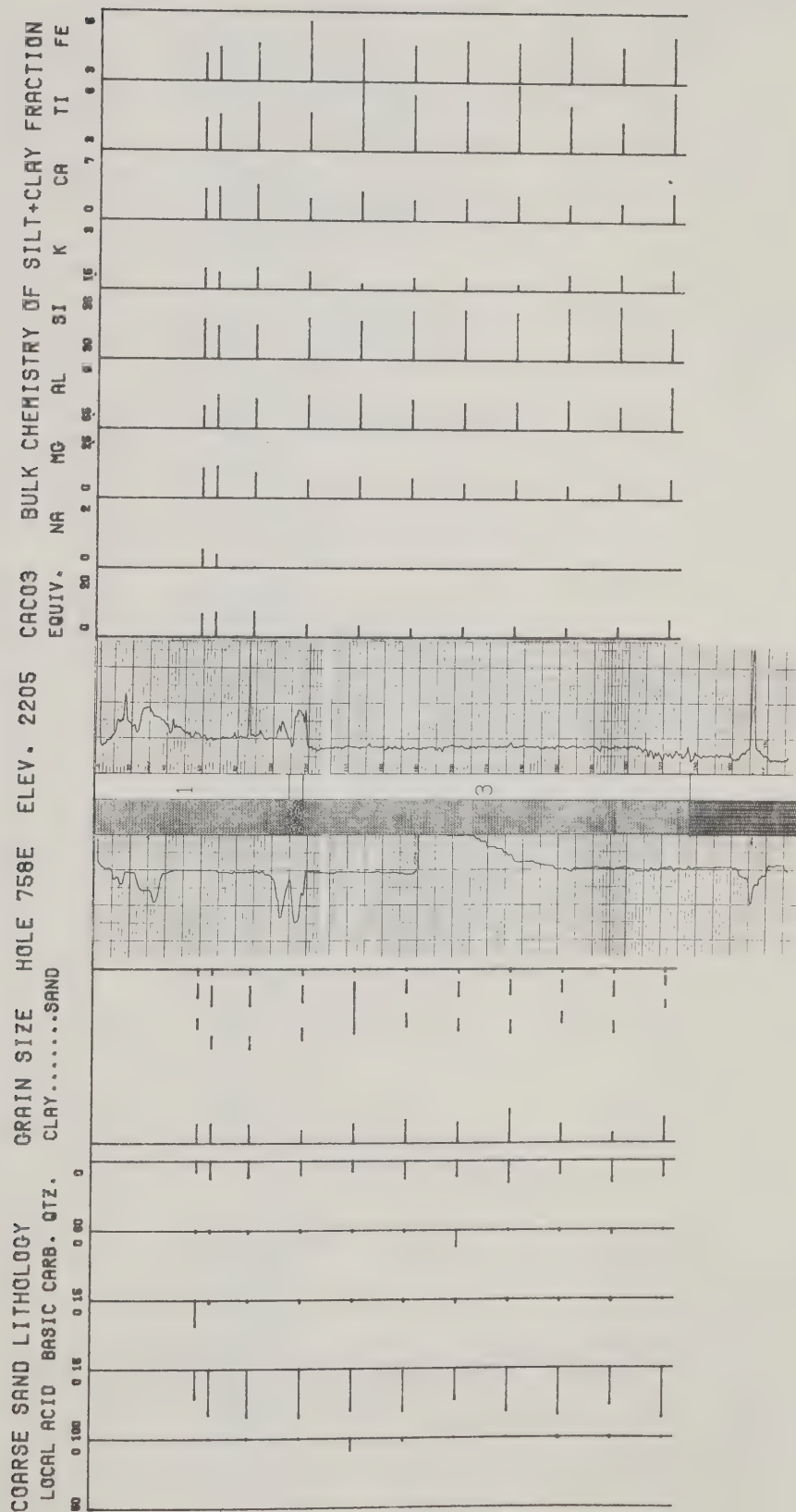






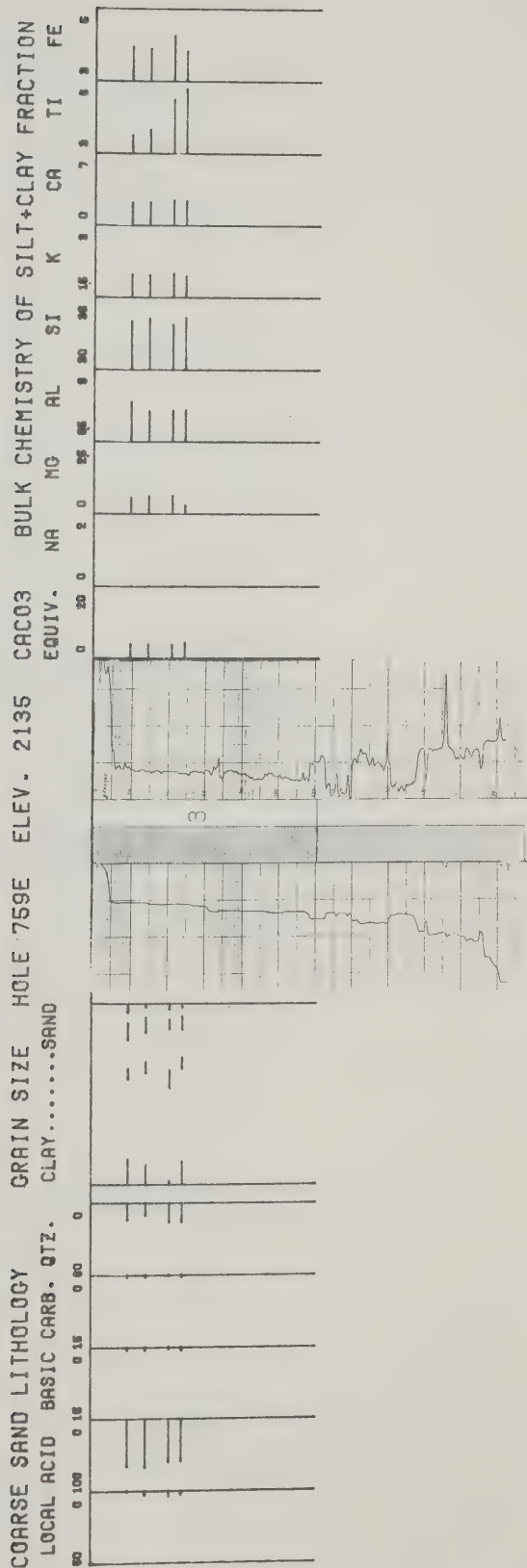




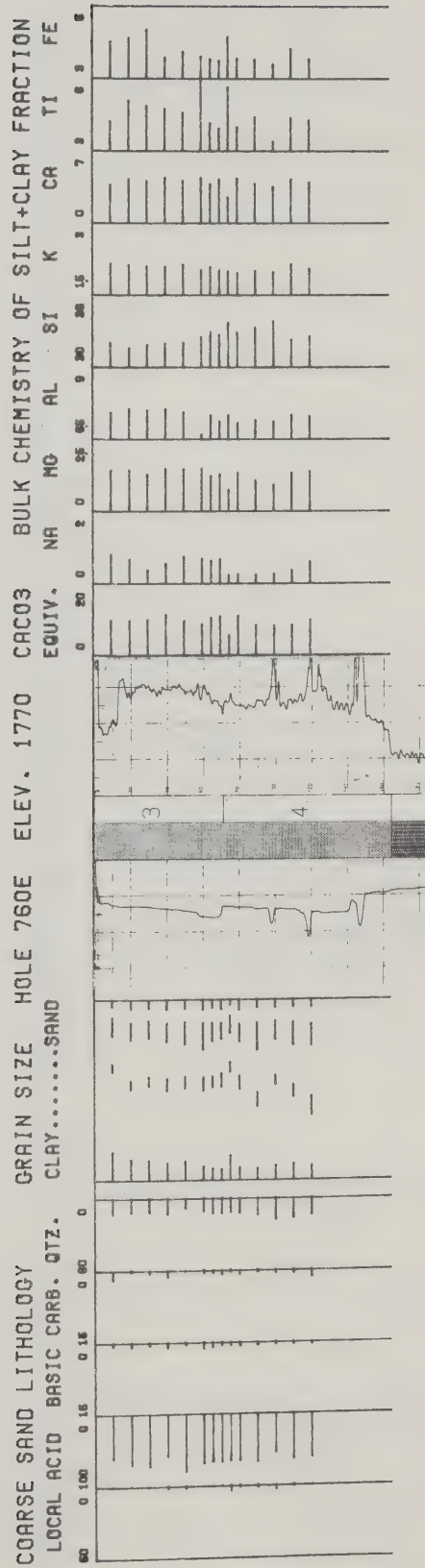




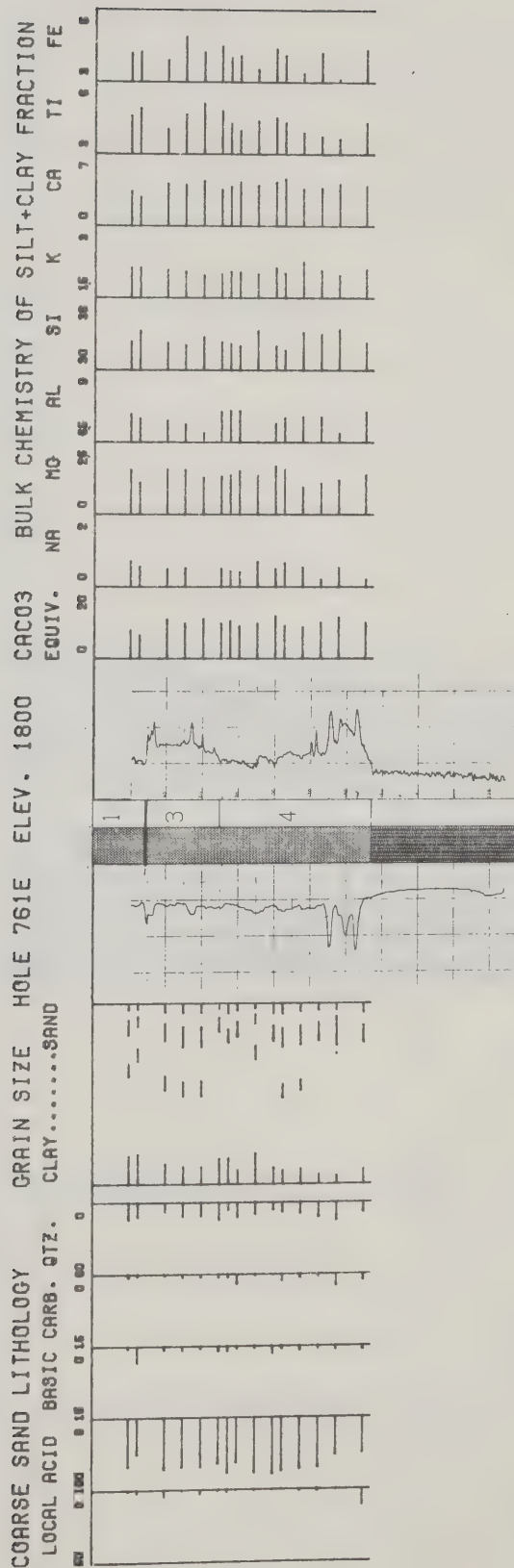




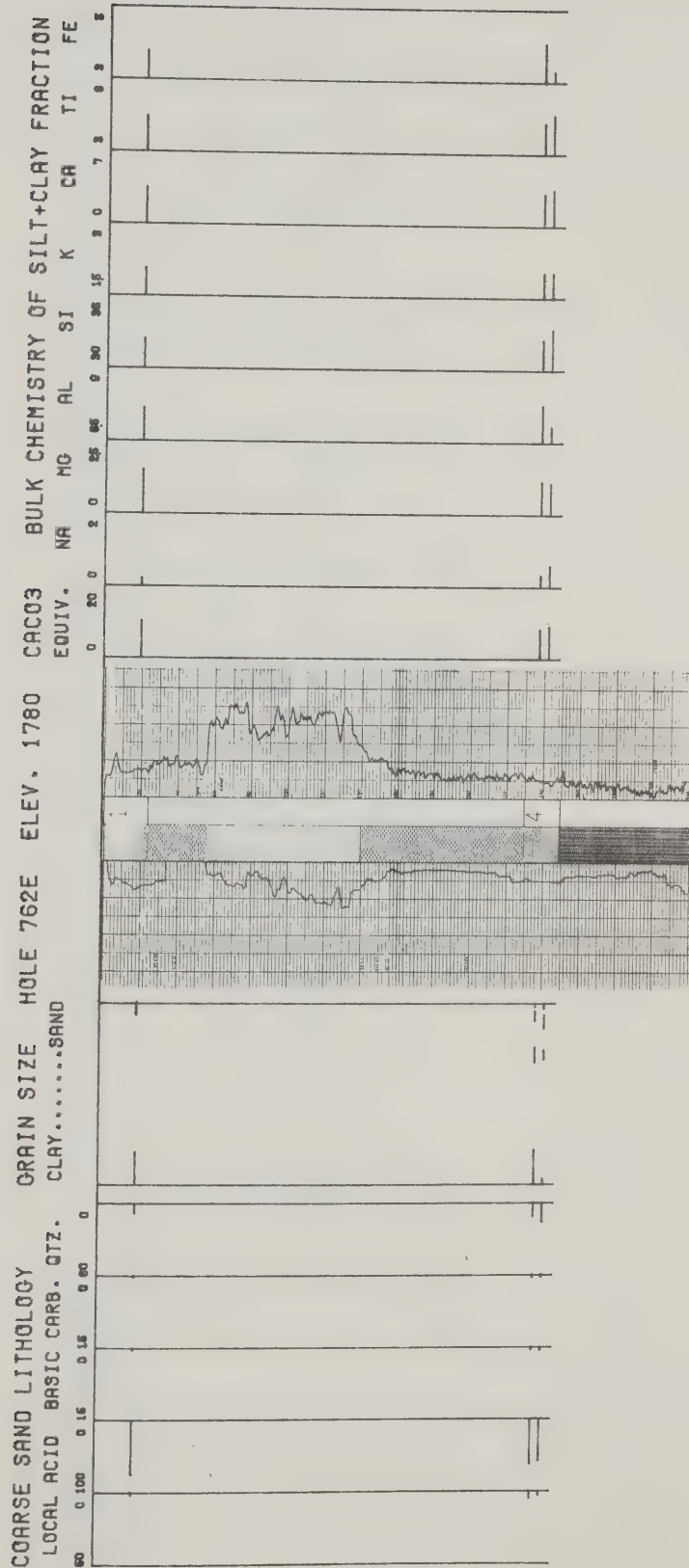






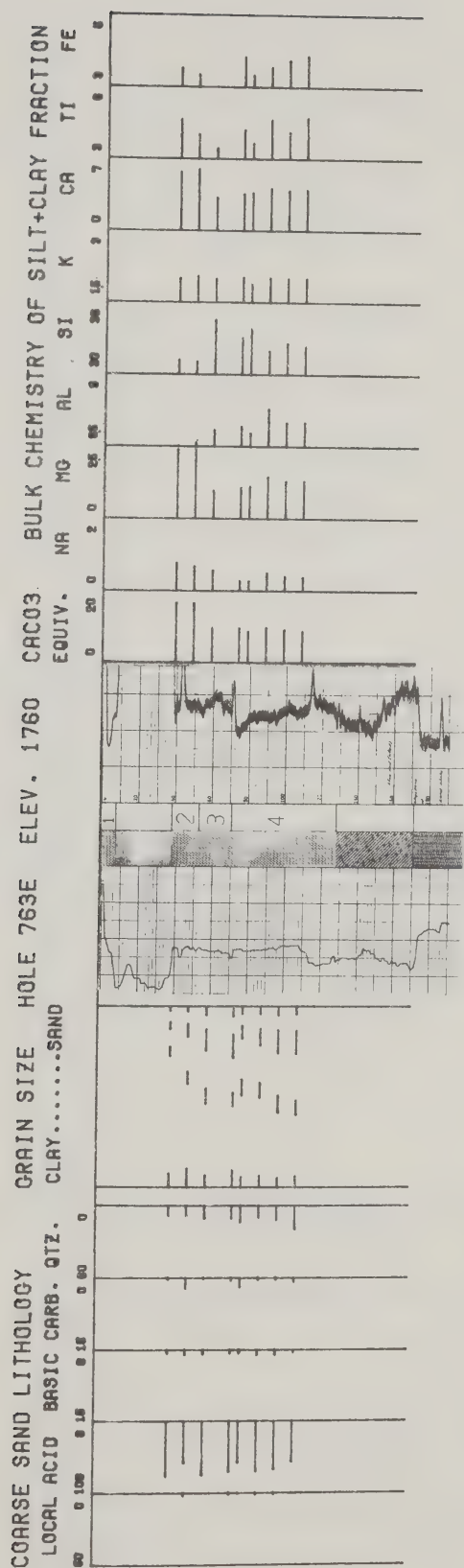




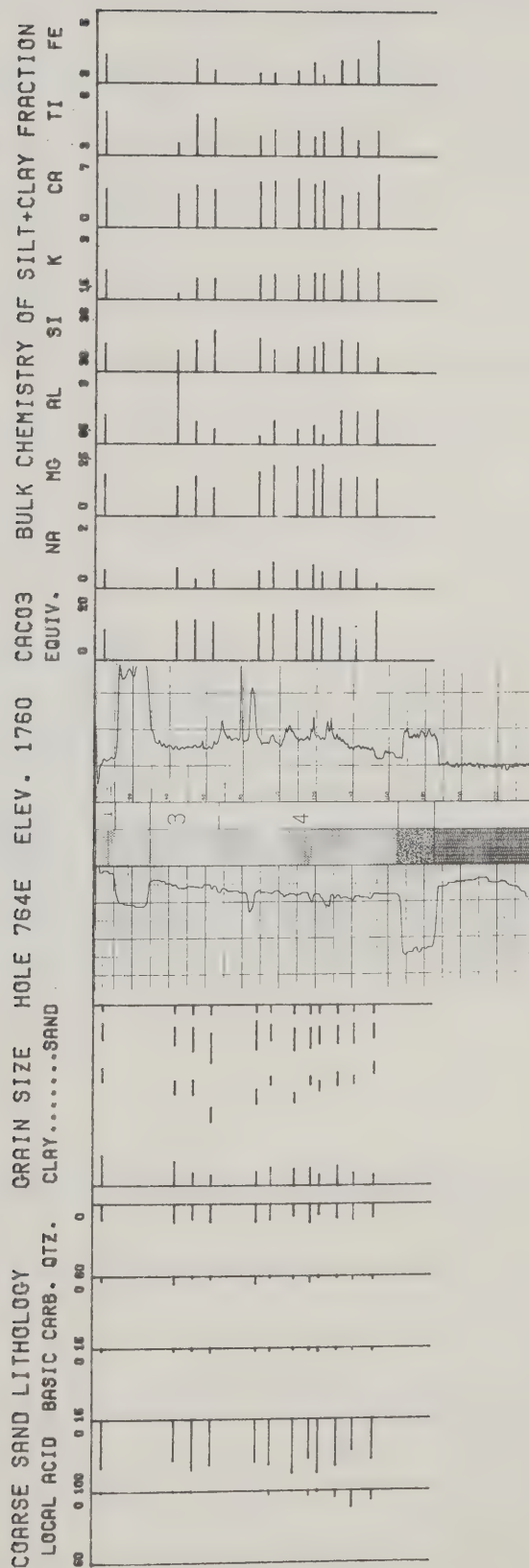




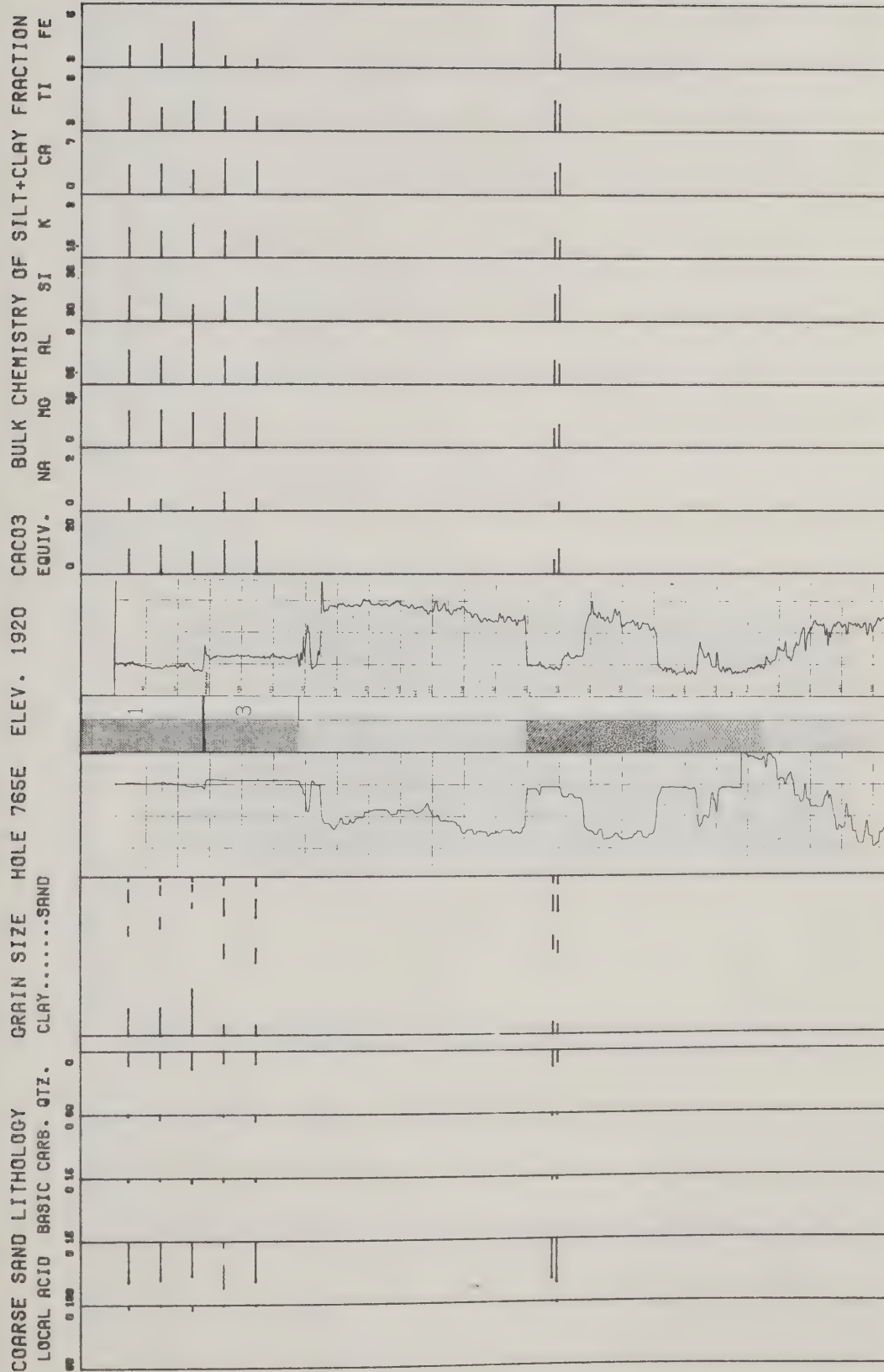




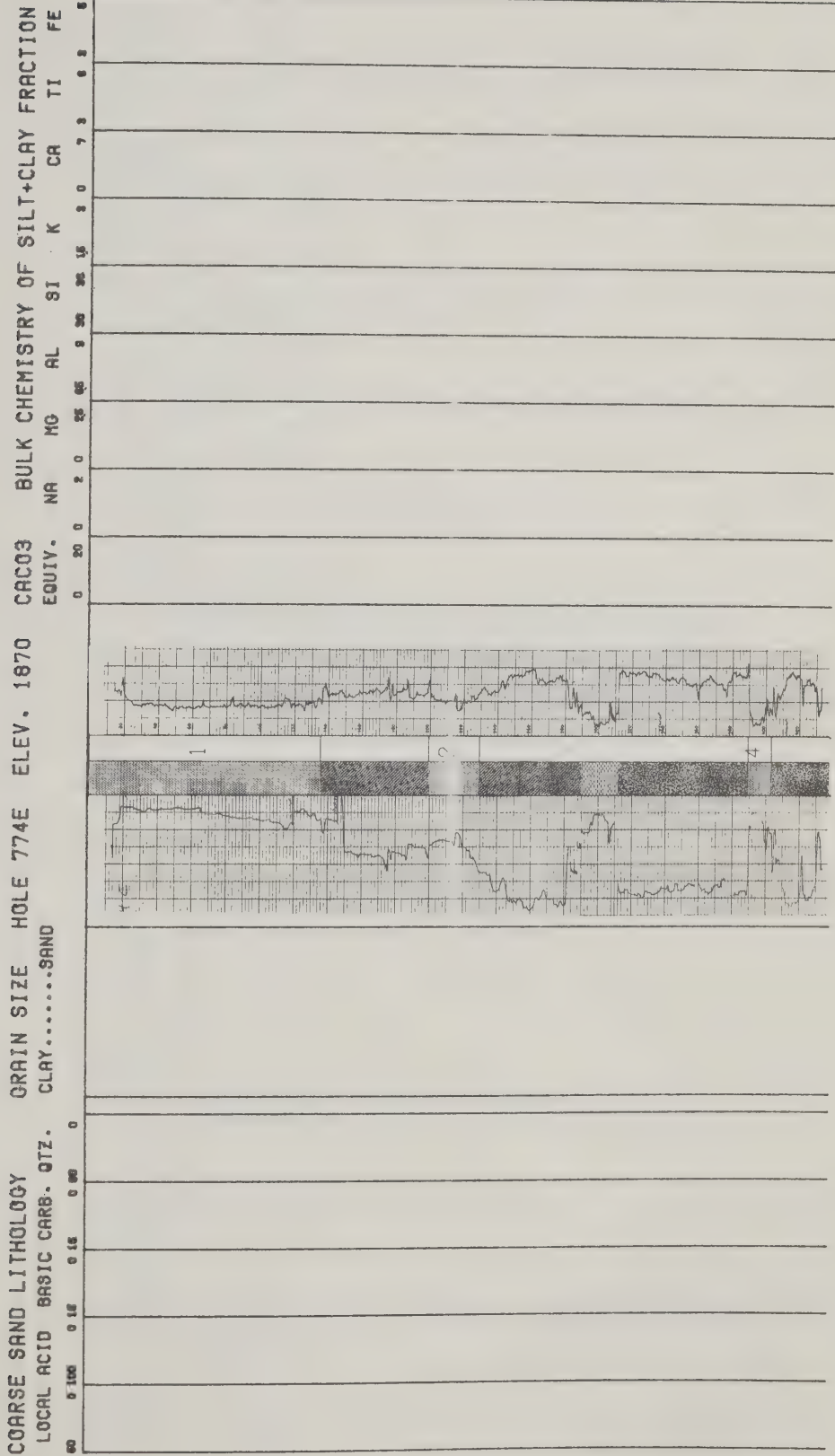






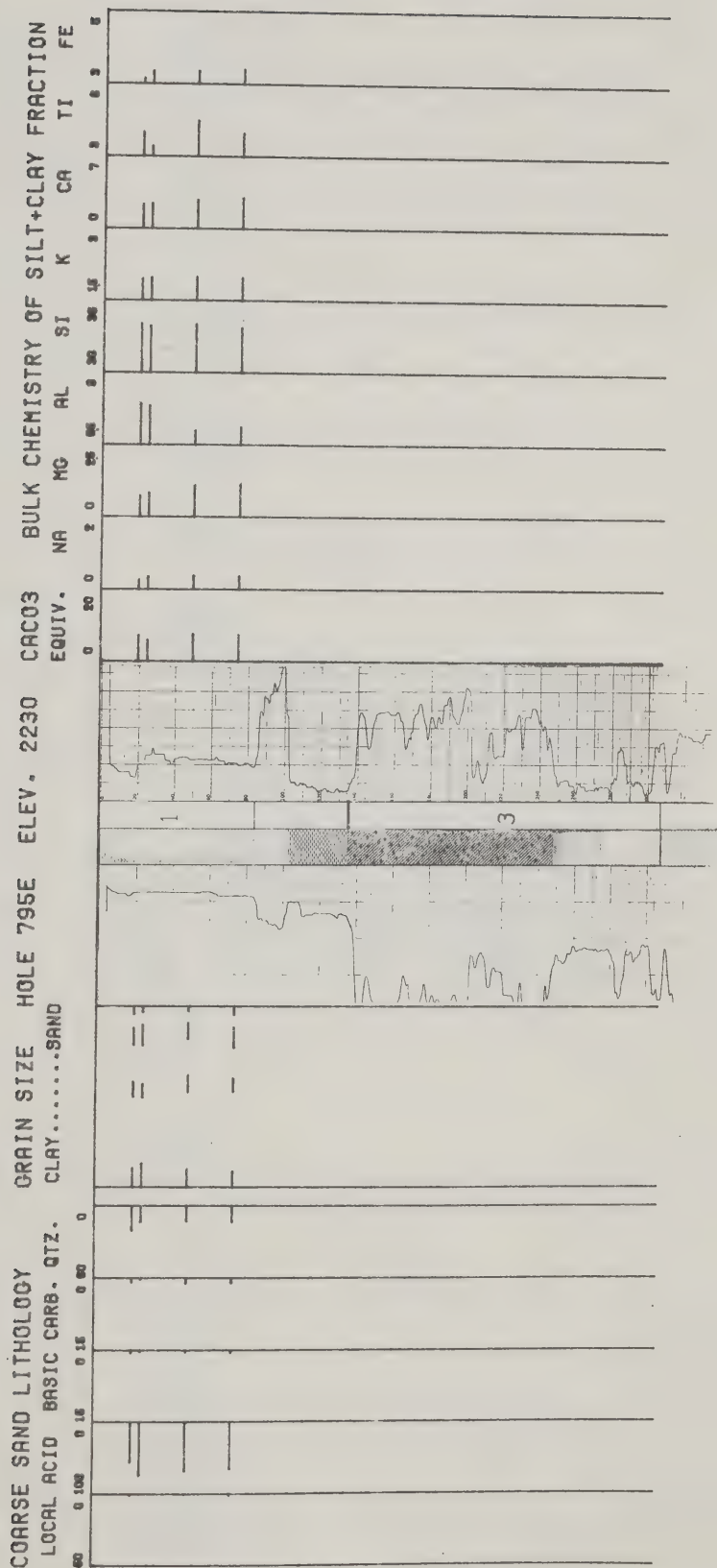




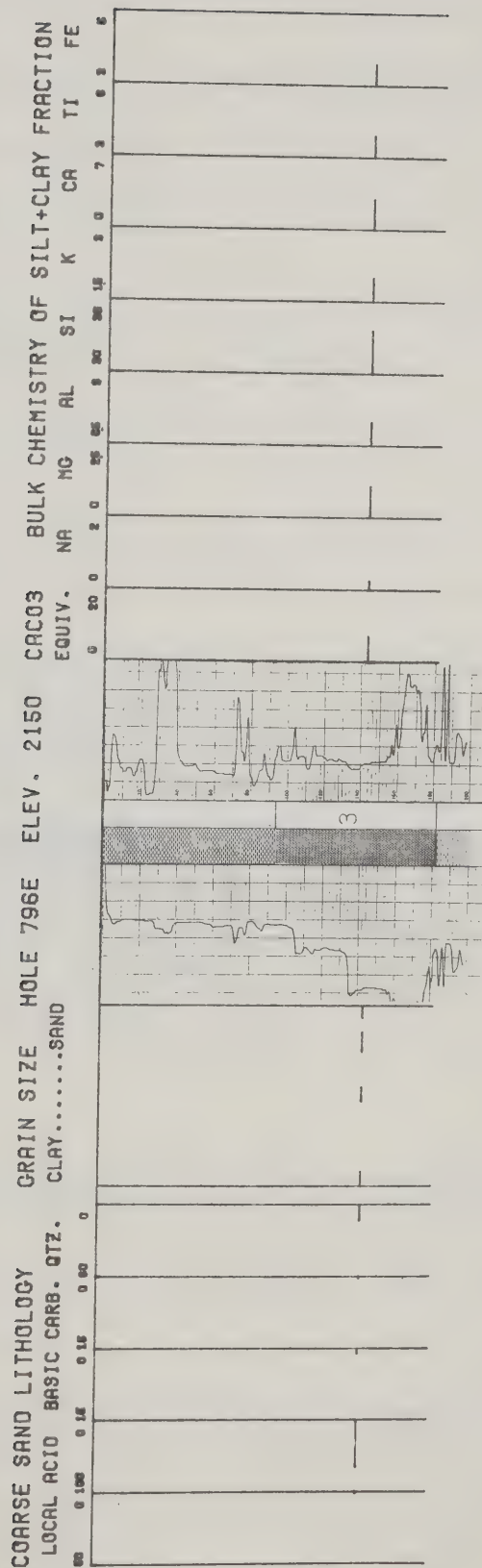










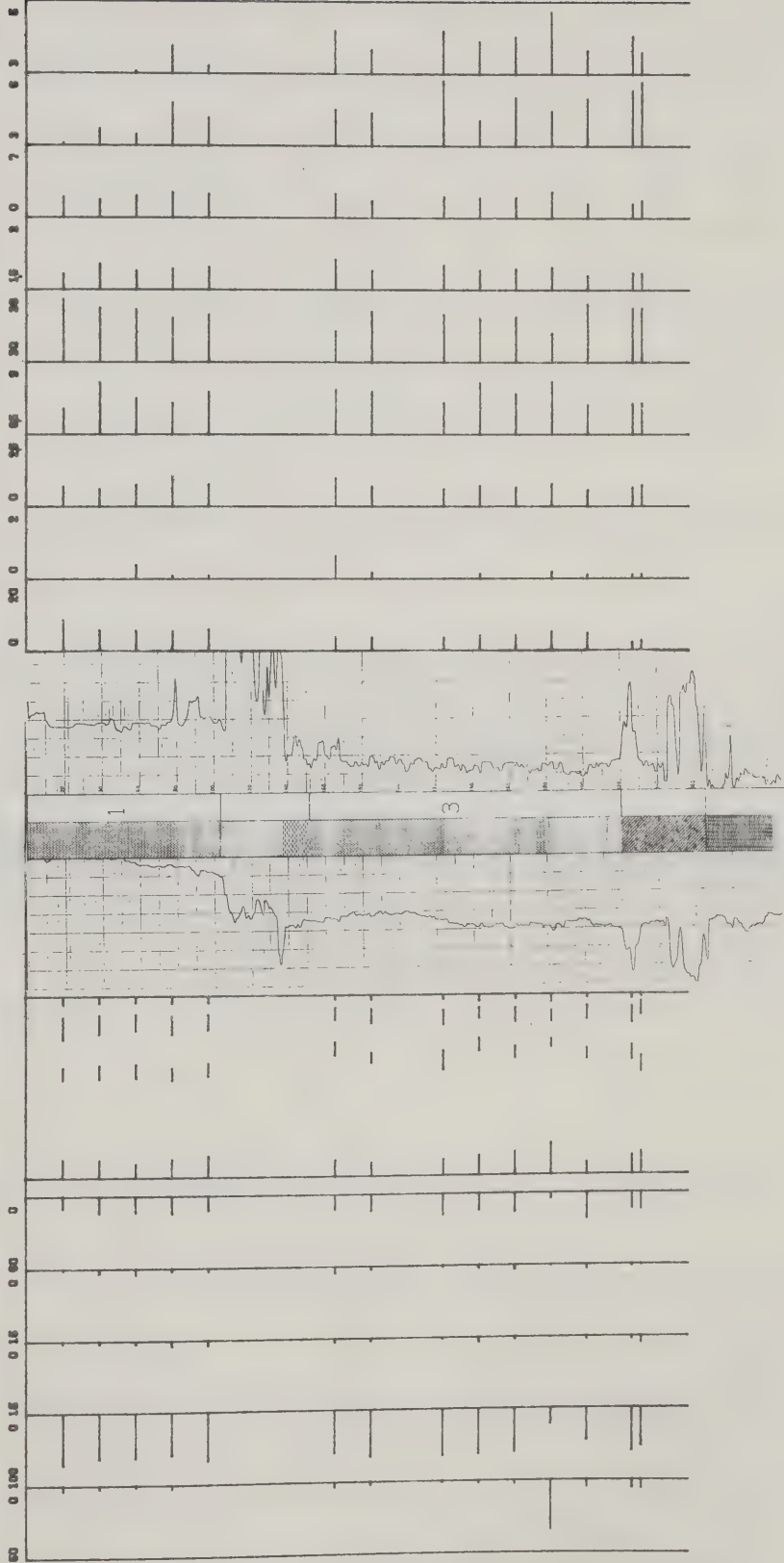




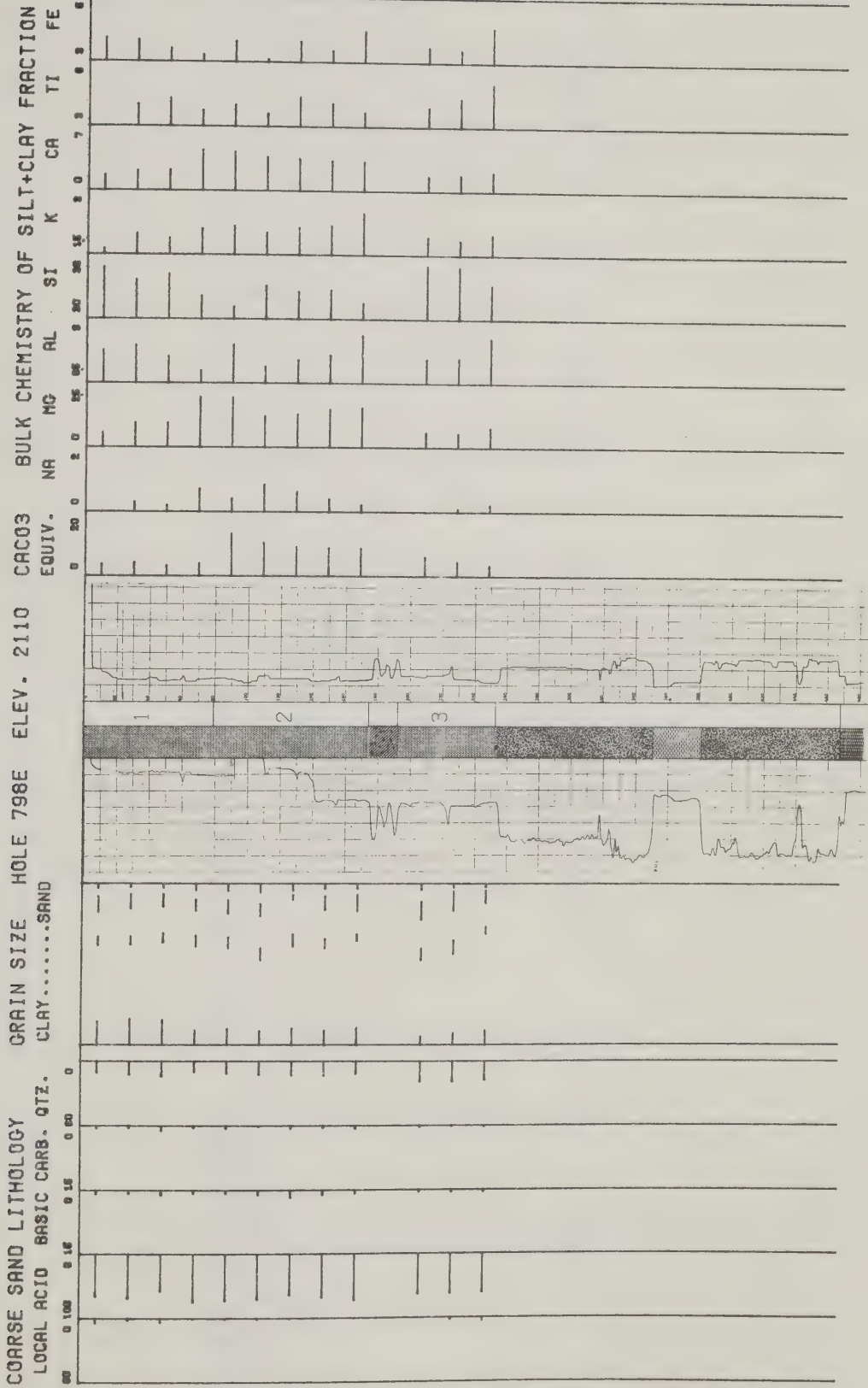
COARSE SAND LITHOLOGY      GRAIN SIZE      HOLE 797E      ELEV. 2105      CAC03      BULK CHEMISTRY OF SILT+CLAY FRACTION

LOCAL ACID      BASIC CARB.      QTZ.      CLAY.....SAND

EQUIV.      NA      MG      AL      SI      K      CA      TI      FE

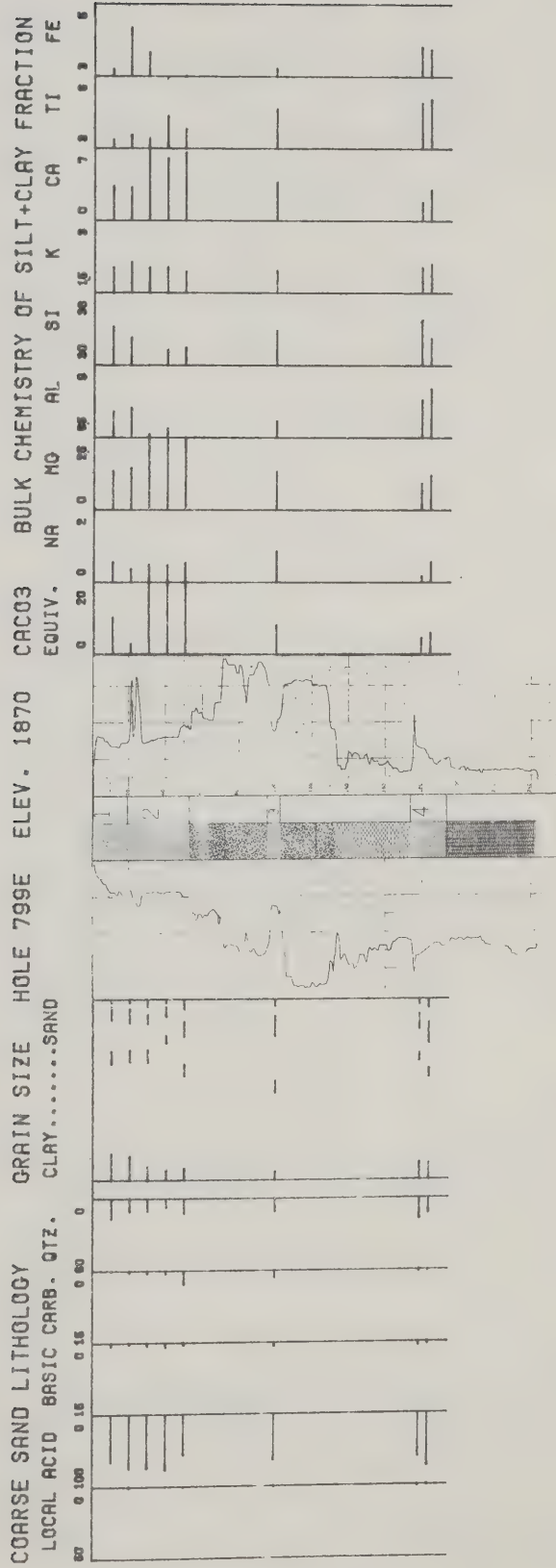




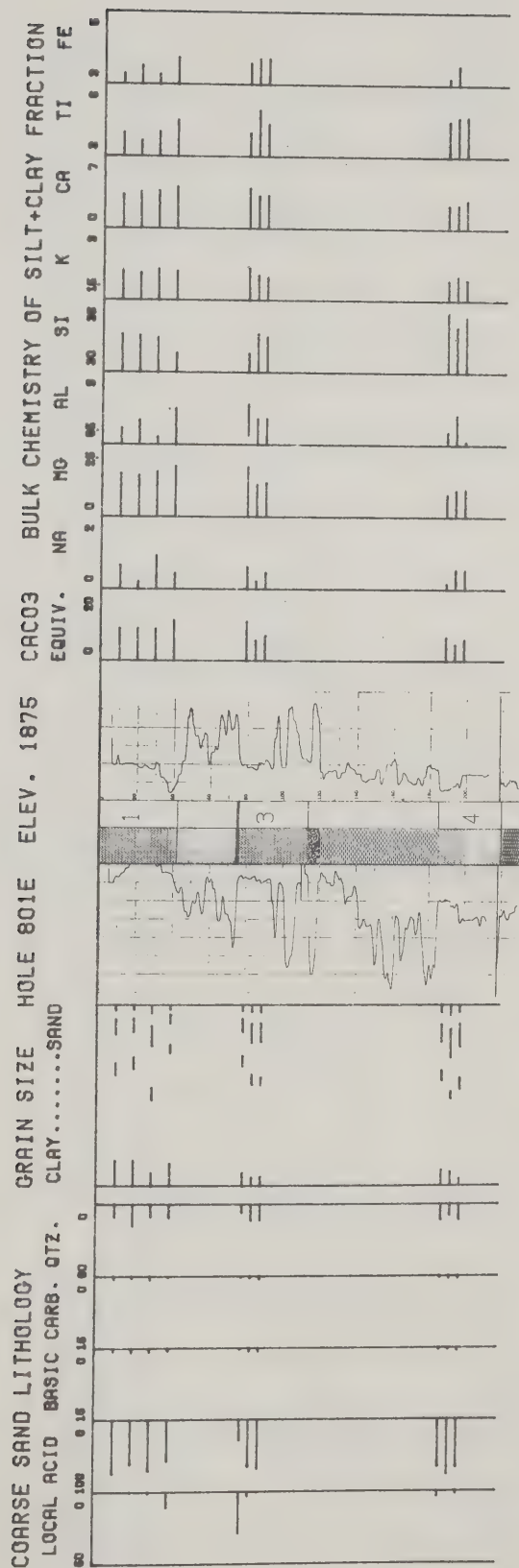




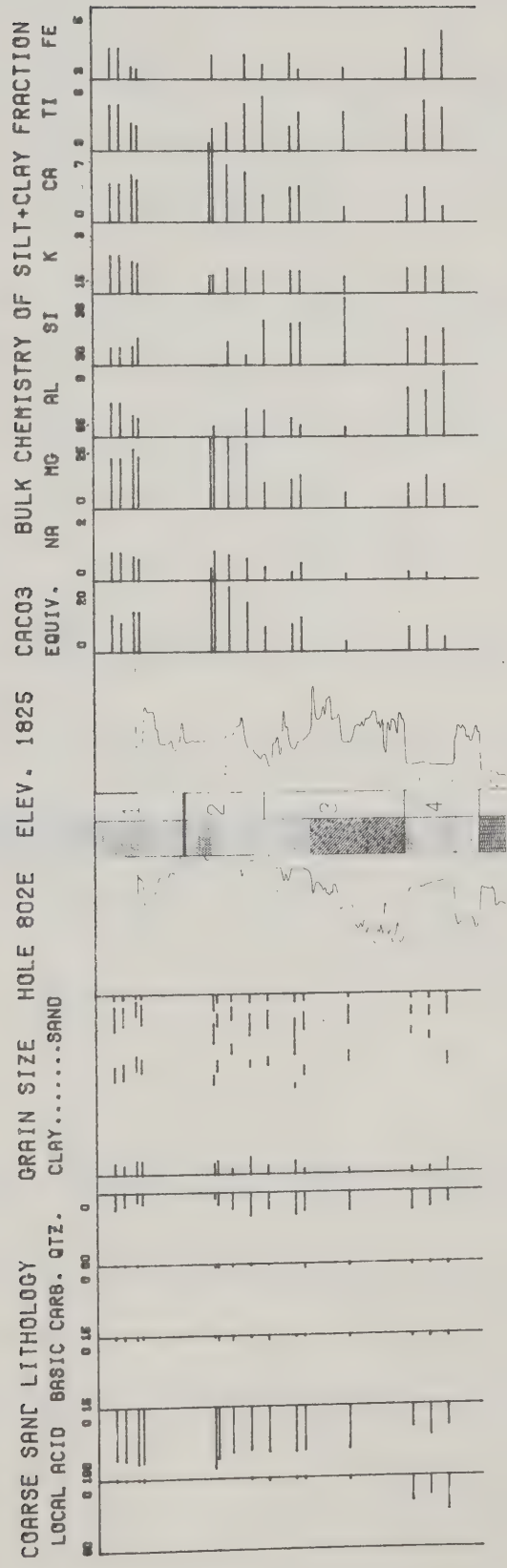




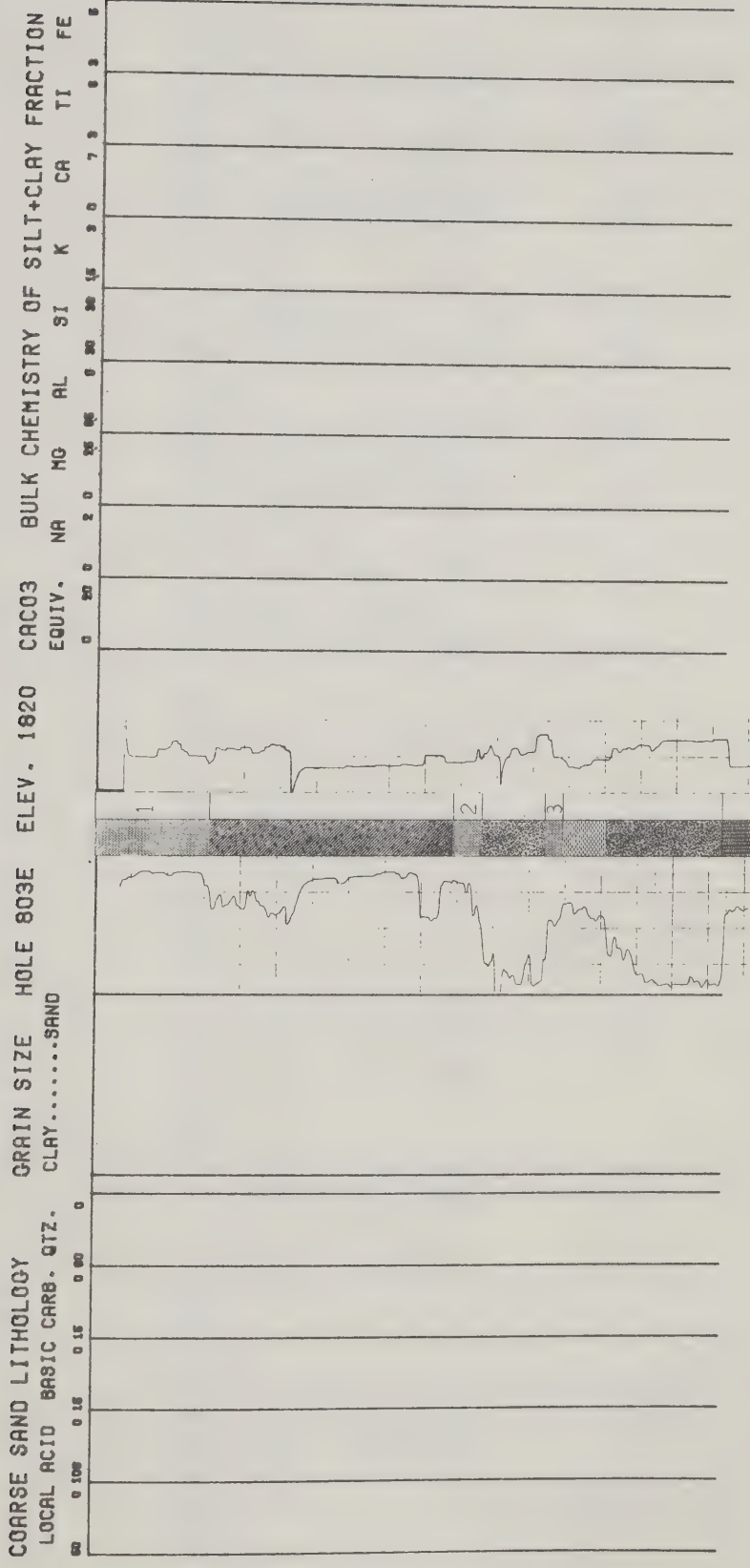






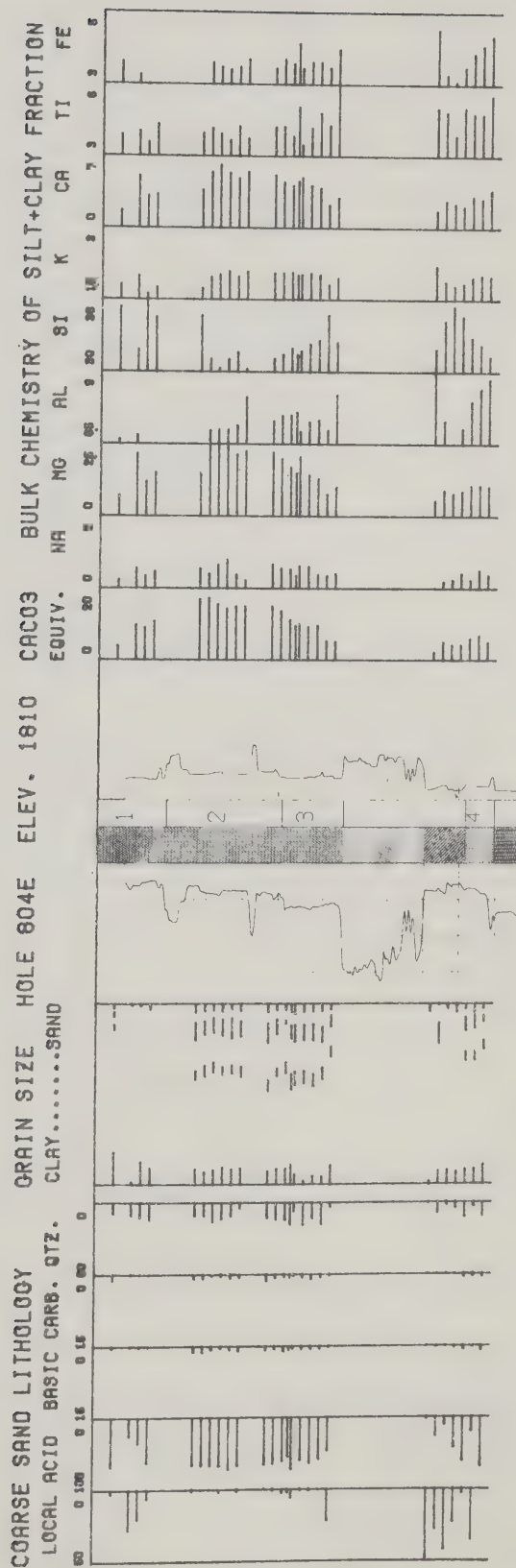




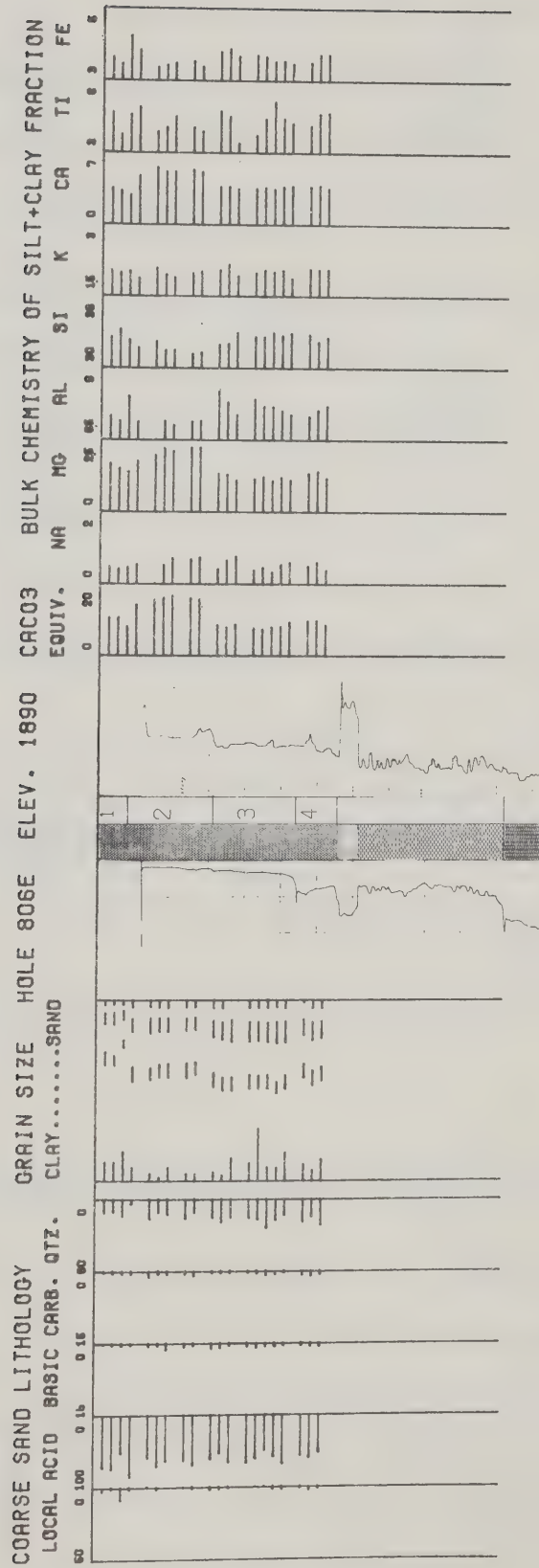




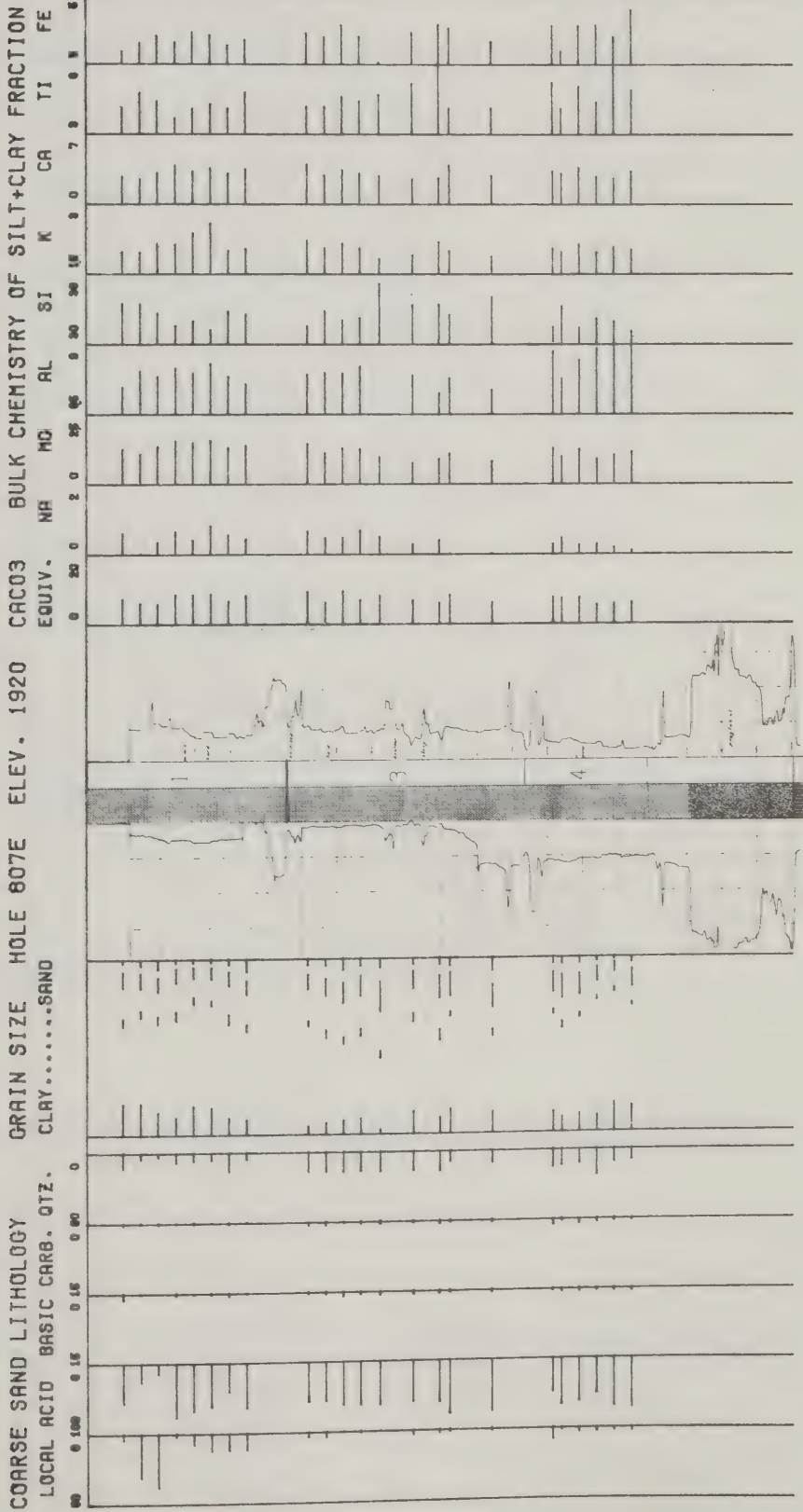




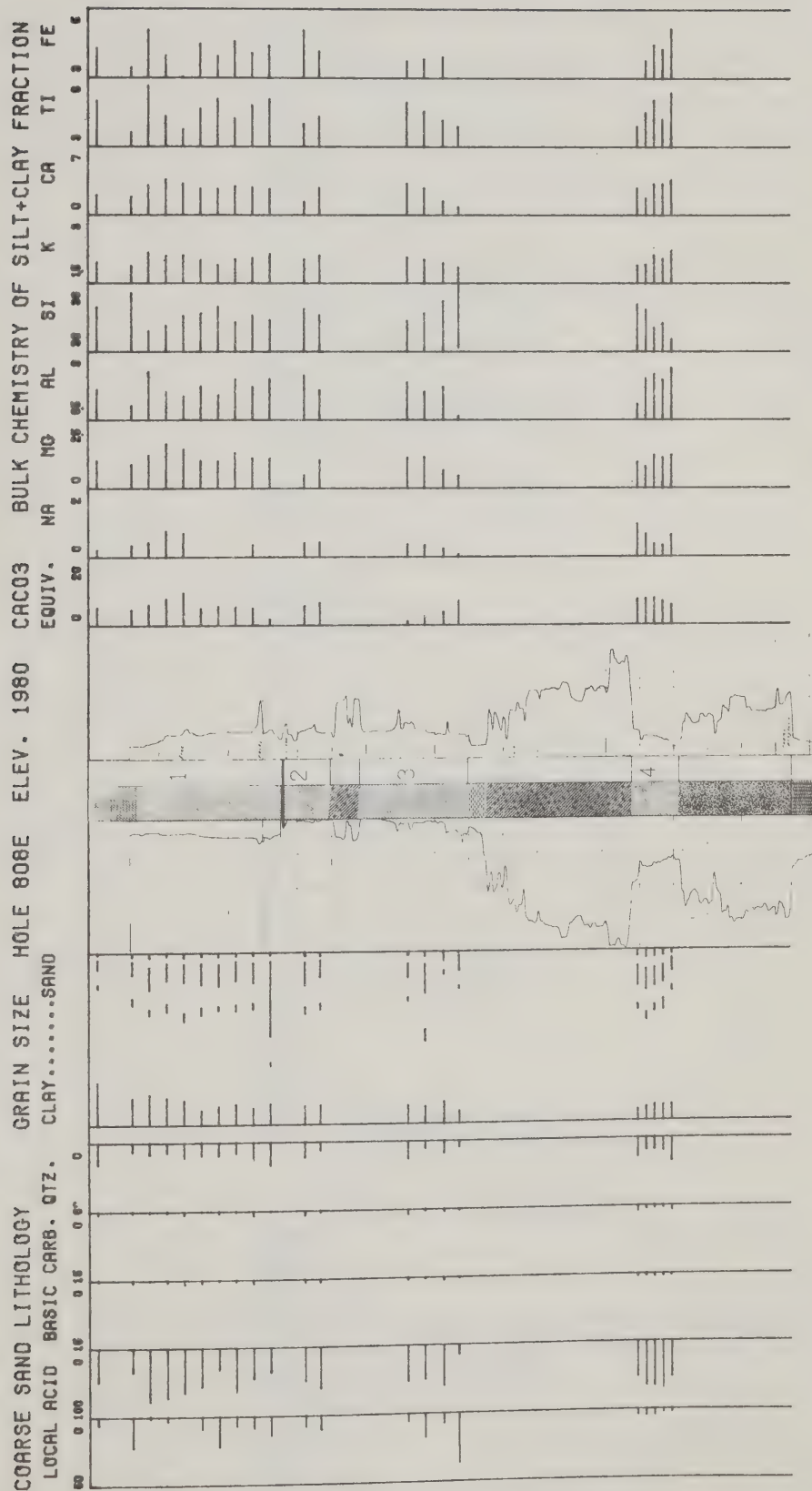






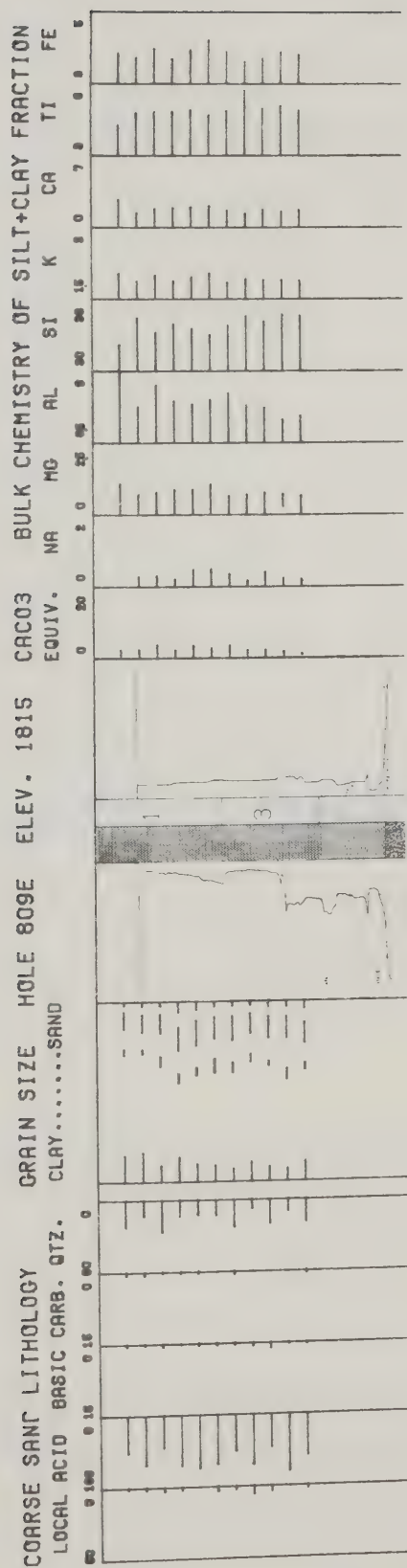








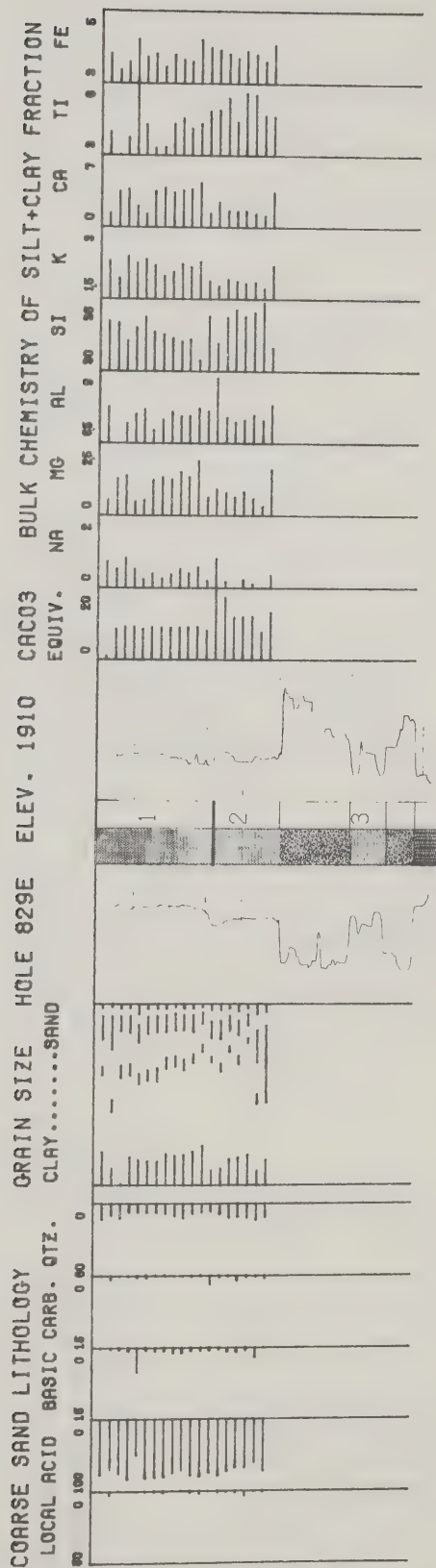




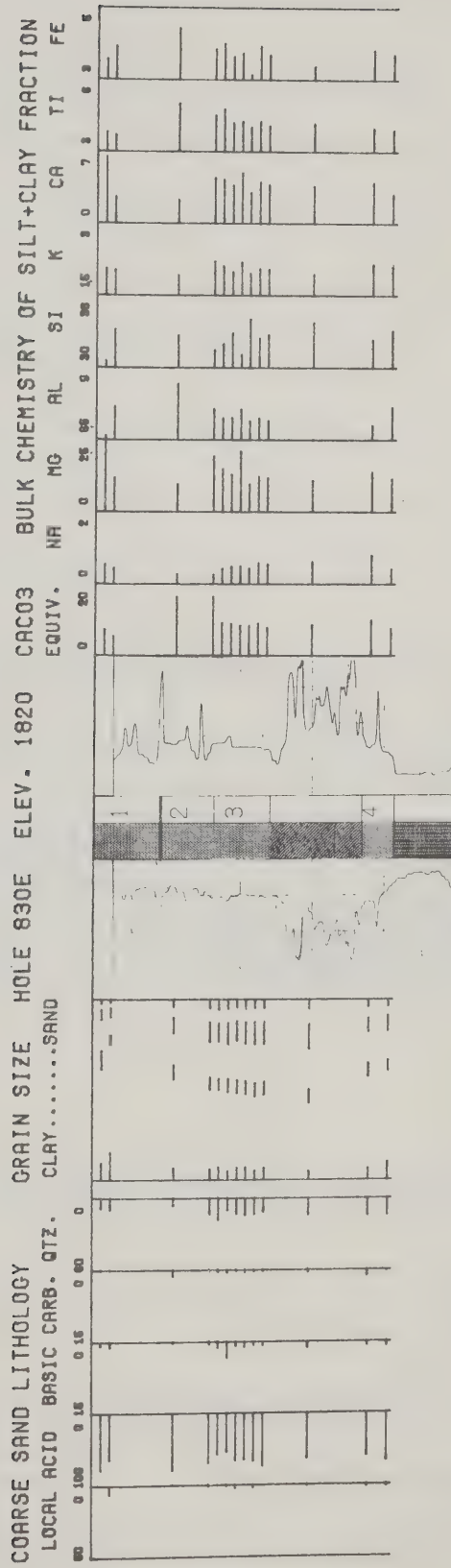






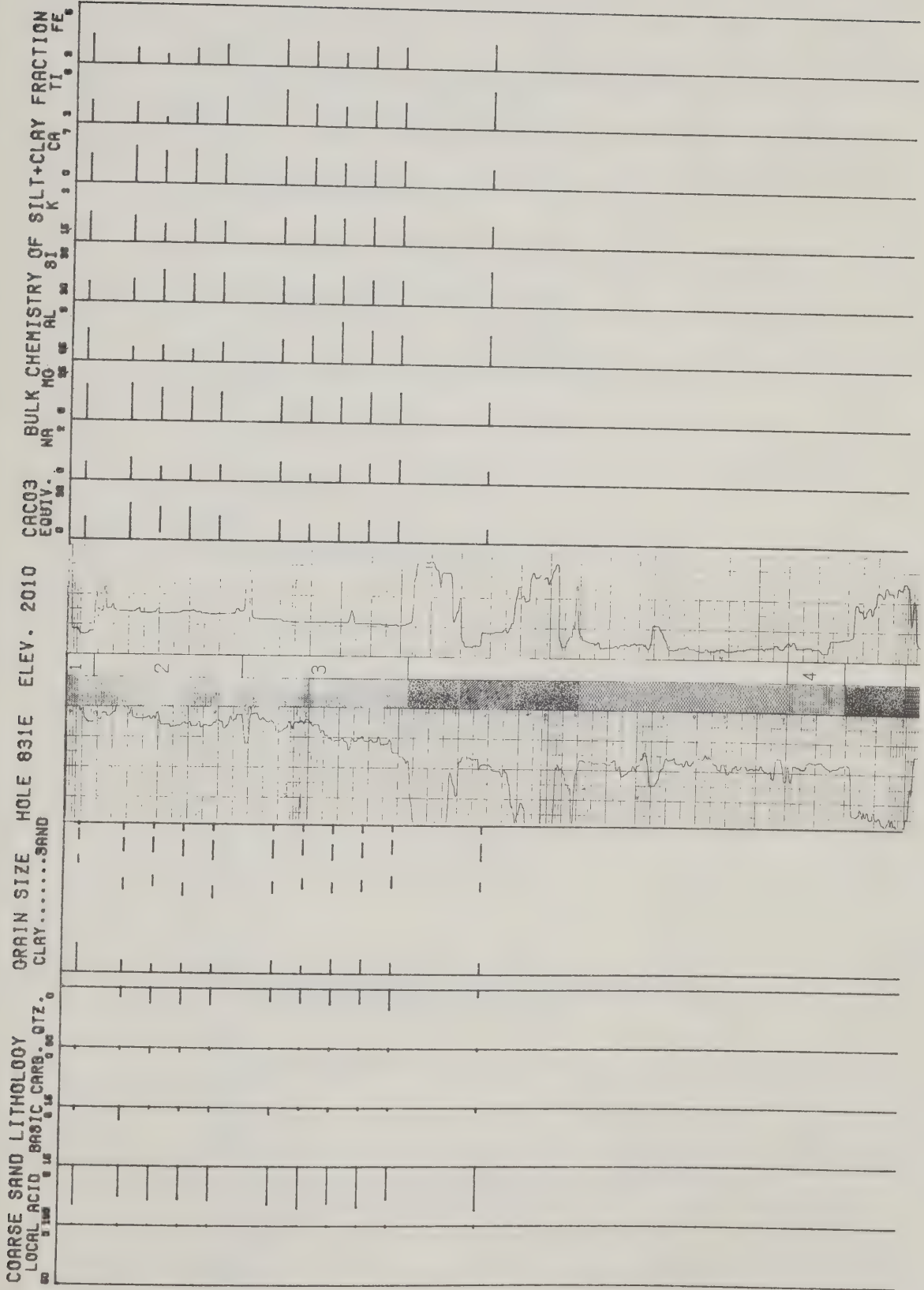




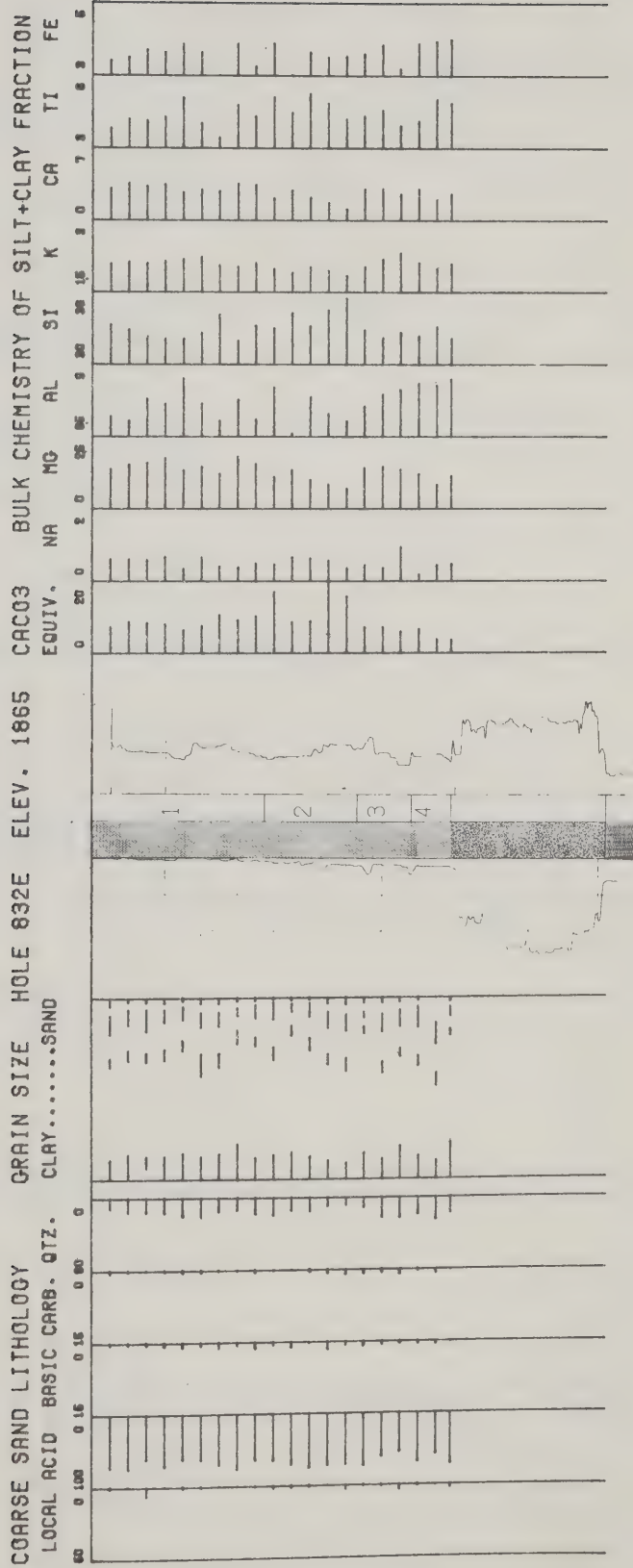




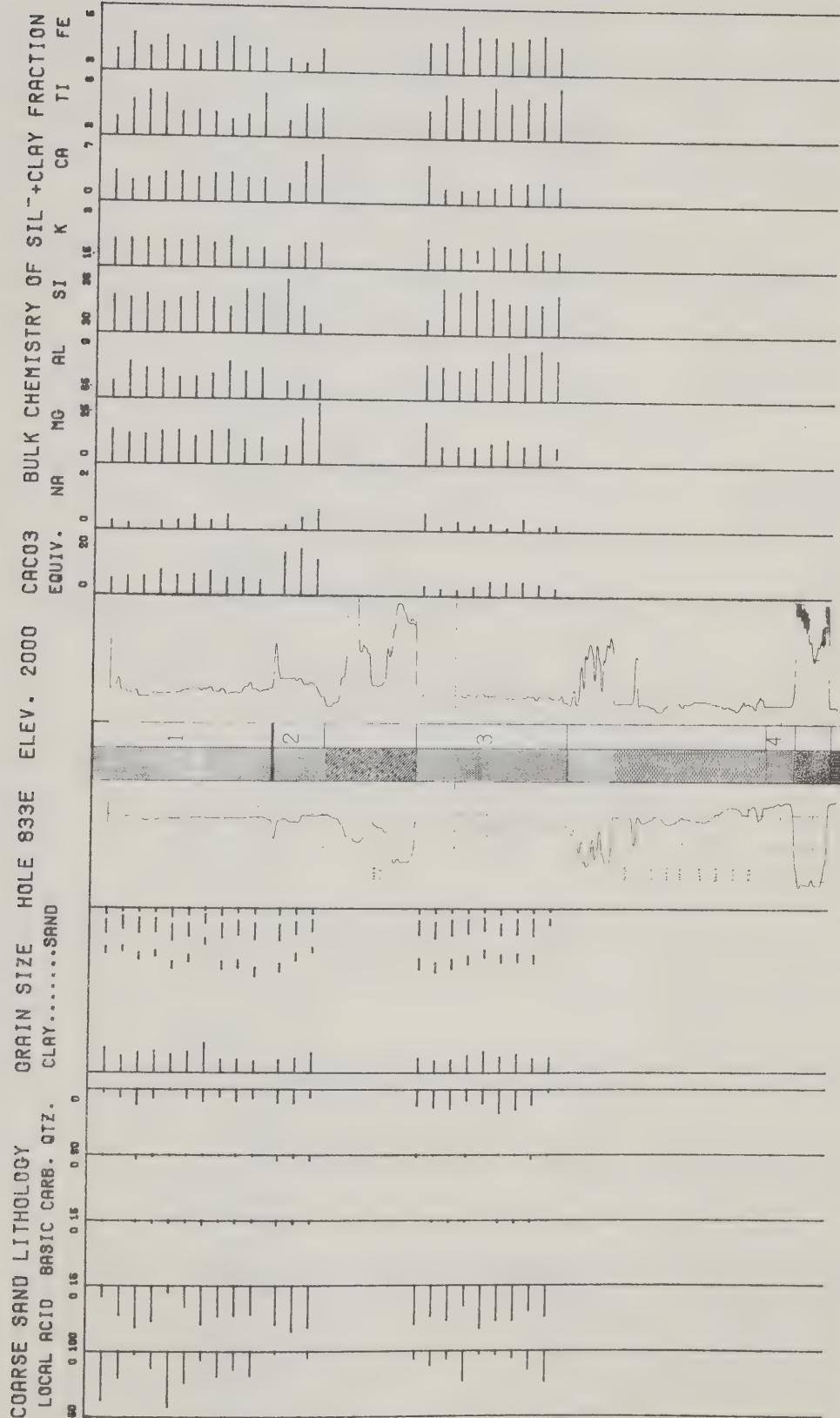




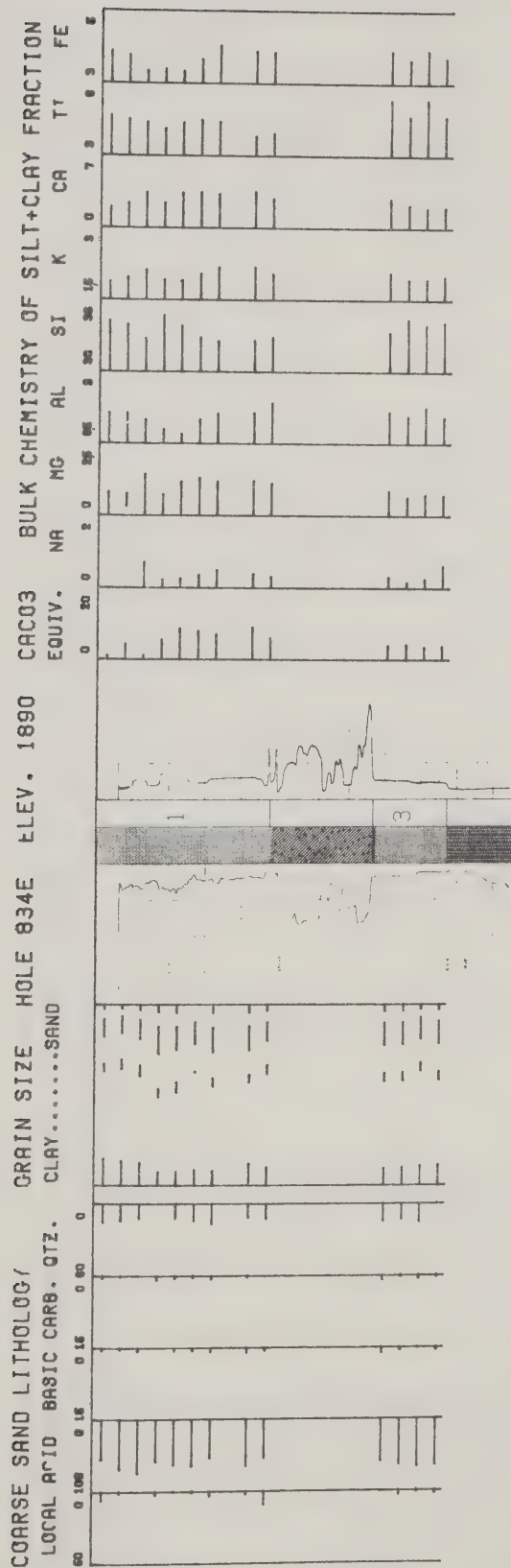






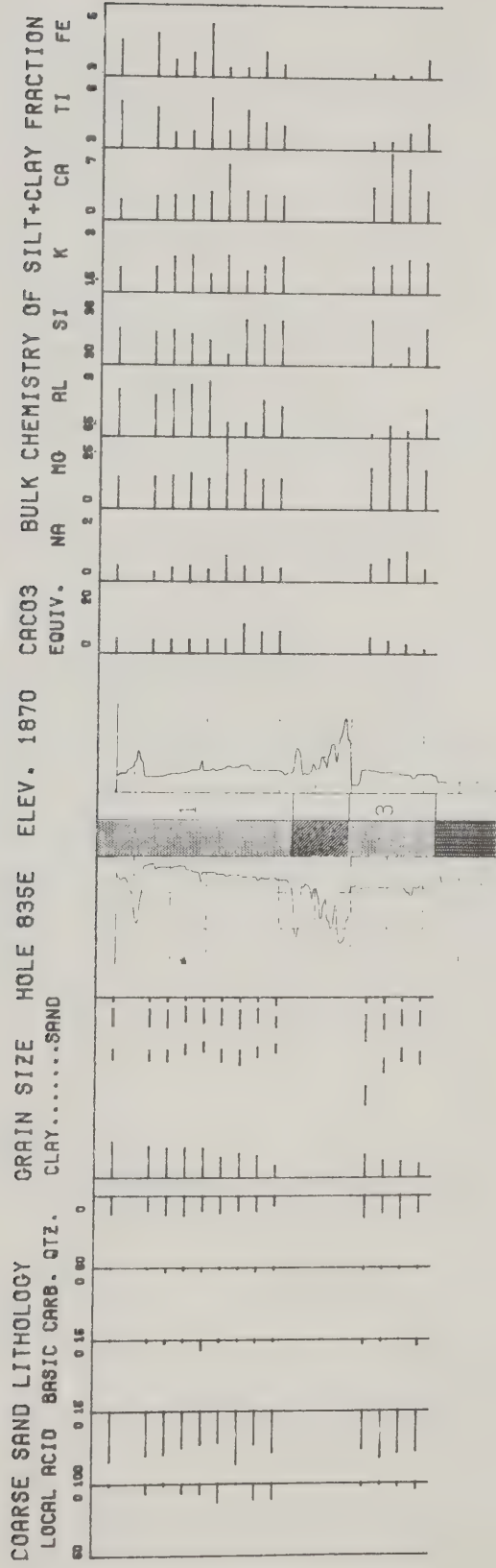




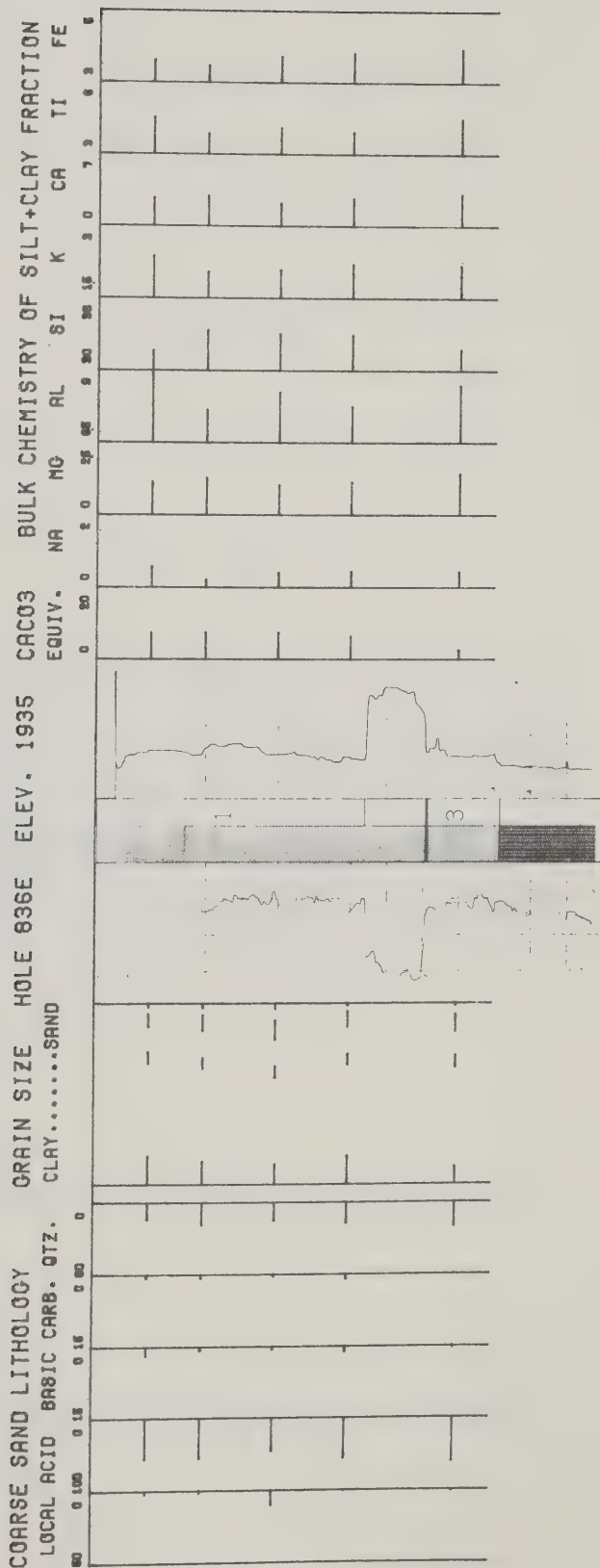




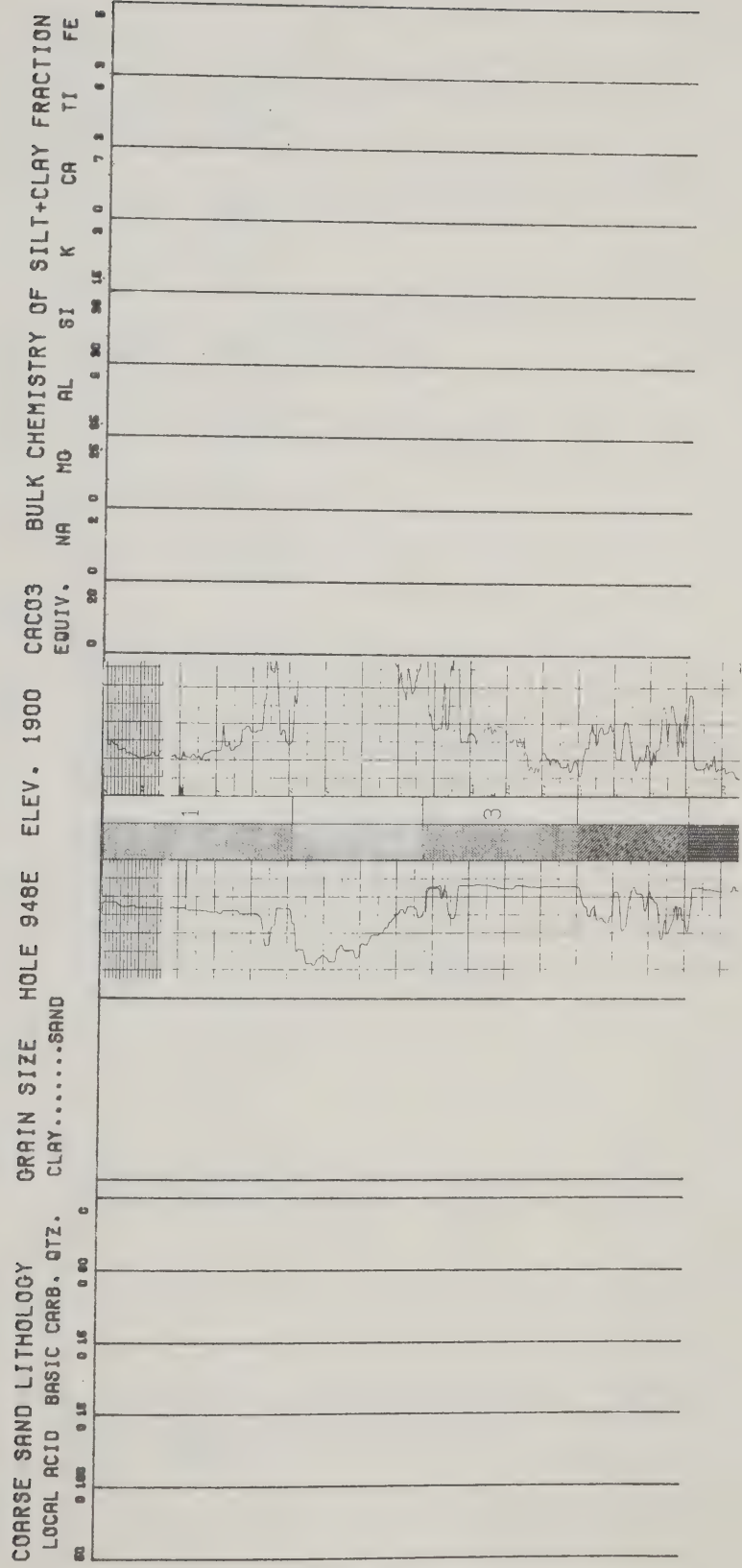




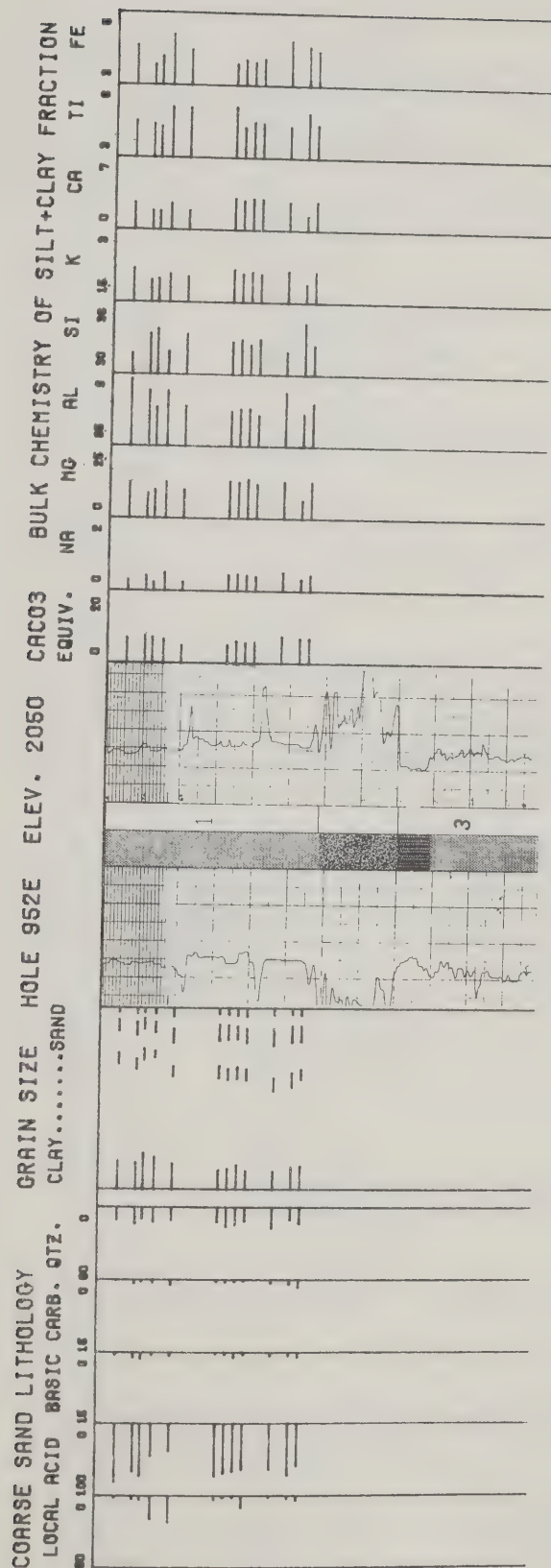






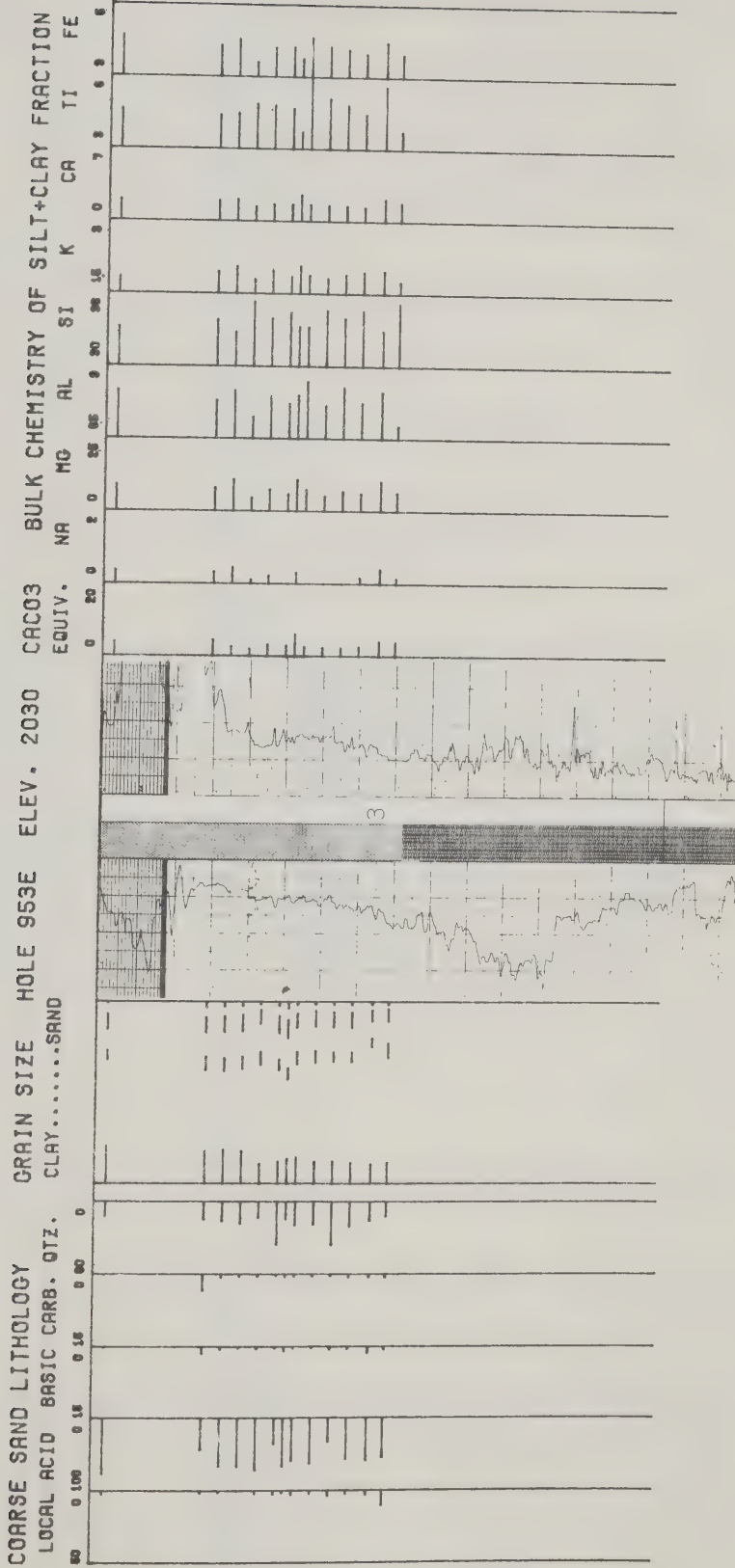




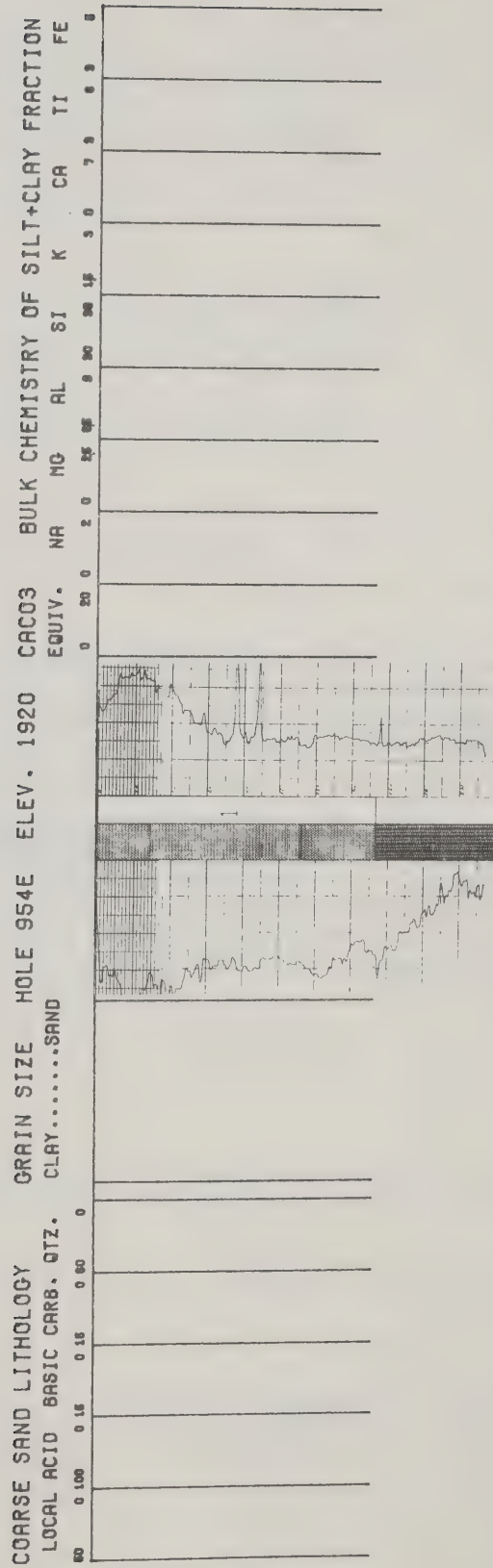




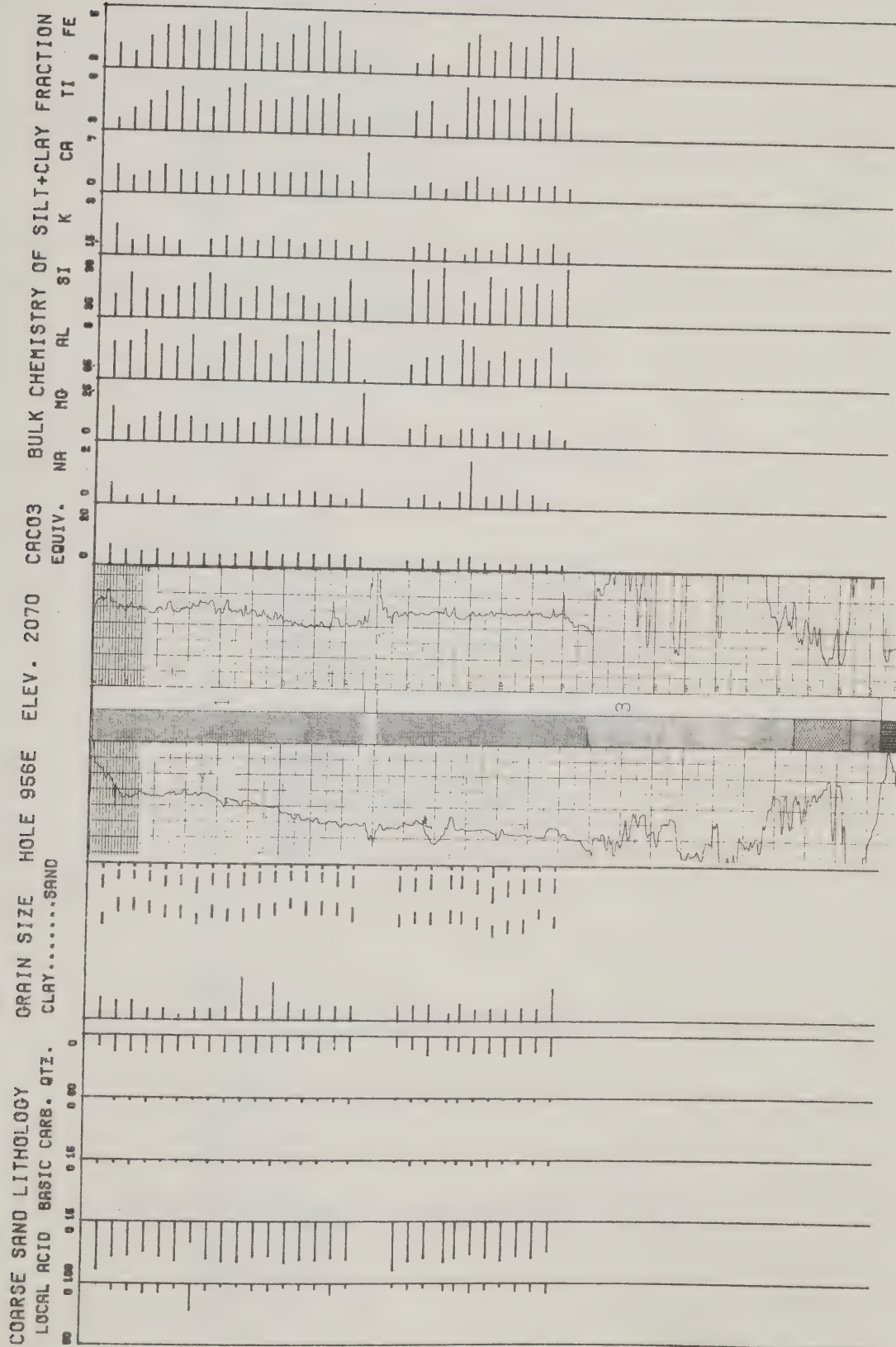




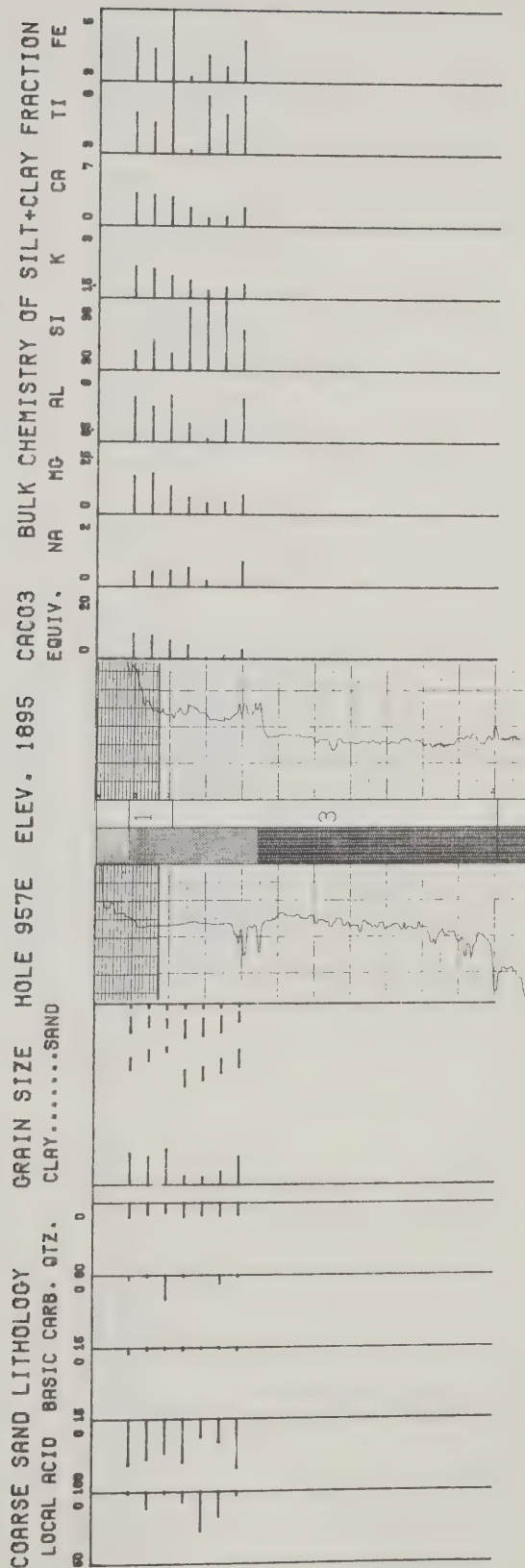






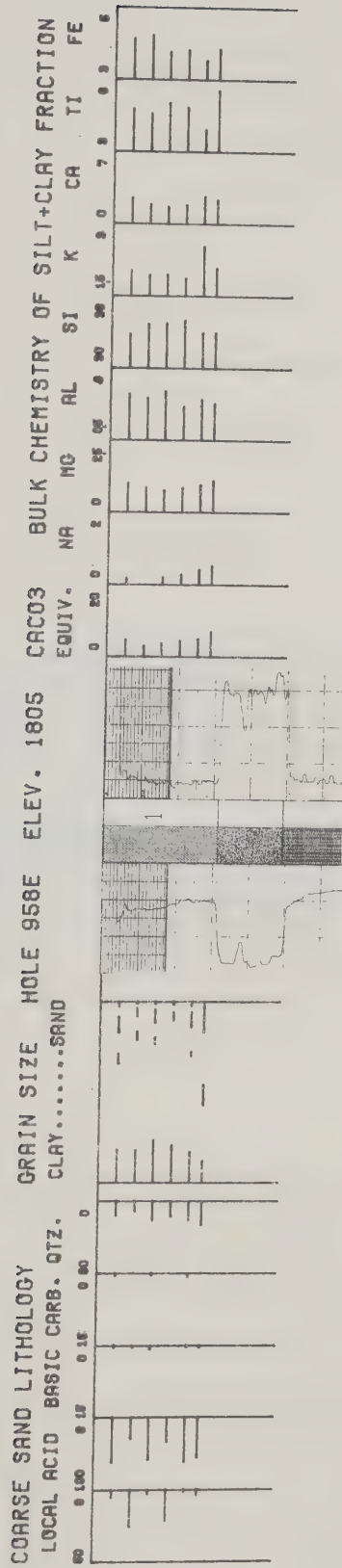




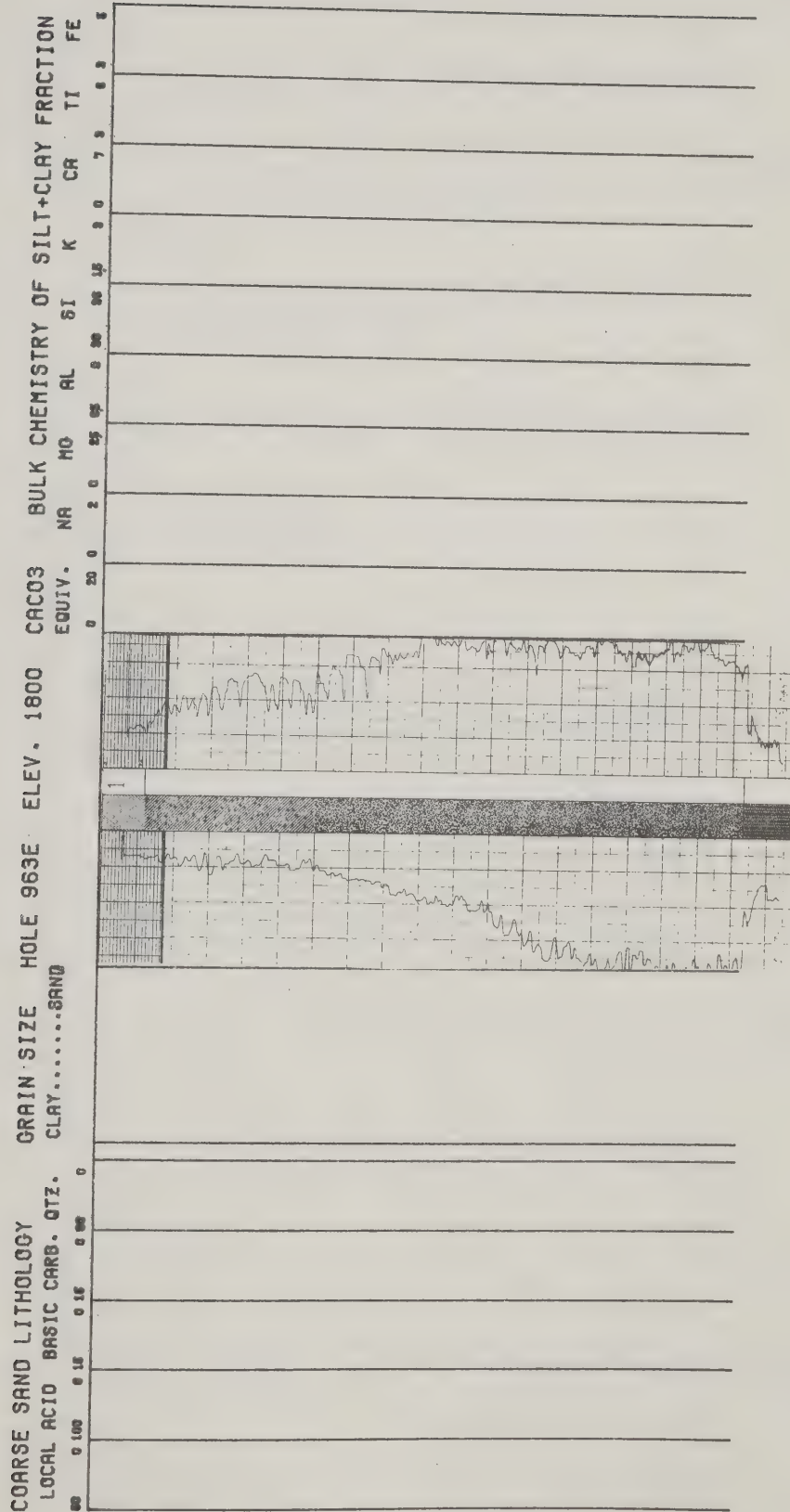




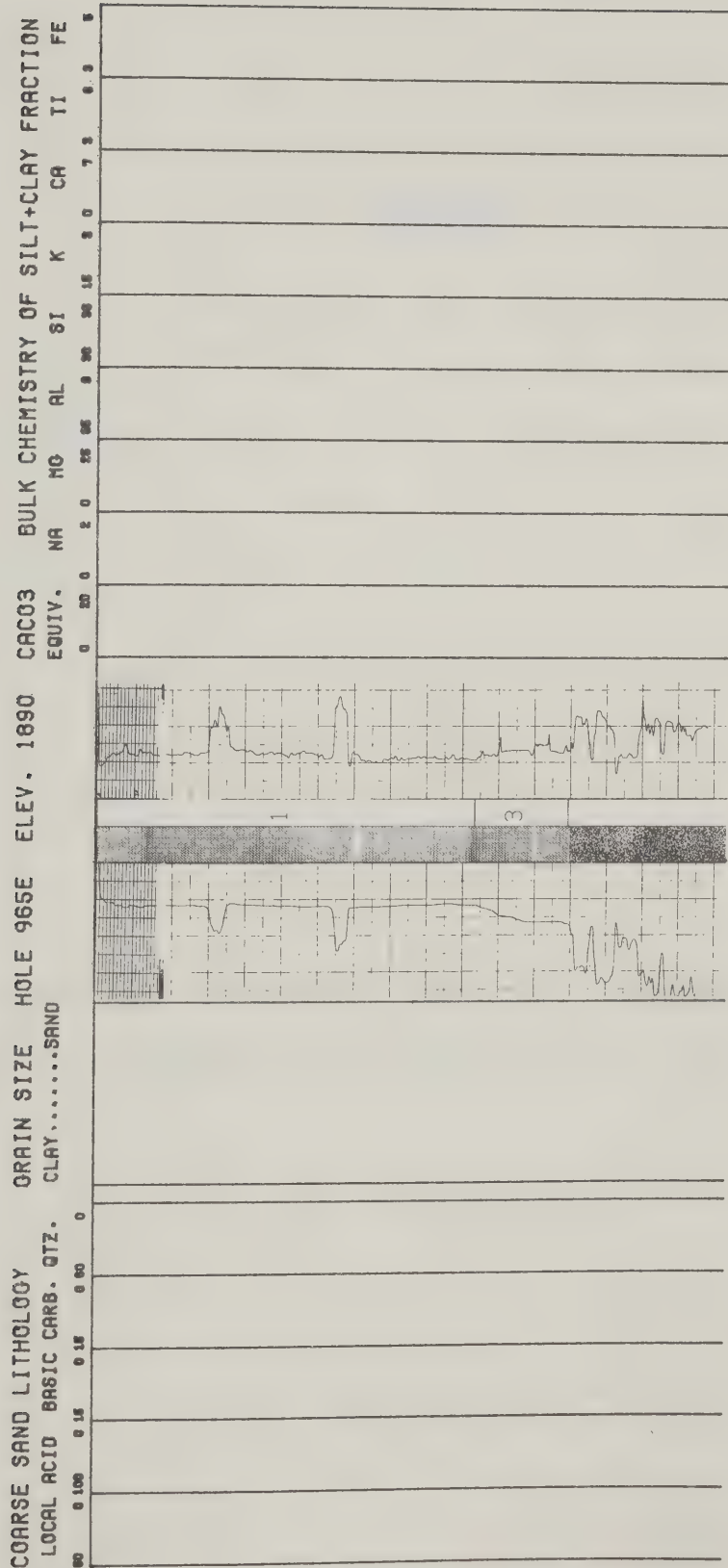














## APPENDIX 5

Analysis of covariance for each analyzed variable,  
comparing between - units and within - units variance.





S1020				
SOURCE OF VARIATION				
COVARIATES				
EAST				
NORTH				
MAIN EFFECTS				
UNIT				
EXPLAINED				
RESIDUAL				
TOTAL				

SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
119.110	2	59.555	28.721	0.000
103.337	1	103.337	49.836	0.000
13.327	1	13.327	6.427	0.011
33.194	3	11.065	5.336	0.001
33.194	3	11.065	5.336	0.001
152.304	5	30.461	14.690	0.000
2017.563	973	2.074		
2169.867	978	2.219		

S2040				
SOURCE OF VARIATION				
COVARIATES				
EAST				
NORTH				
MAIN EFFECTS				
UNIT				
EXPLAINED				
RESIDUAL				
TOTAL				

SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
112.989	2	56.494	15.721	0.000
48.752	1	48.752	12.454	0.000
64.748	1	64.748	18.018	0.000
12.796	3	4.265	1.187	0.314
12.796	3	4.265	1.187	0.314
125.785	5	25.157	7.001	0.000
3496.442	973	3.593		
3622.227	978	3.704		



S4060				
SOURCE OF VARIATION				
	SUM OF SQUARES	DF	MEAN SQUARE	SIGNIF. F OF F
COVARIATES				
EAST	128.393	2	64.196	6.603 0.001
NORTH	95.037	1	95.037	9.775 0.002
	29.886	1	29.886	3.074 0.080
MAIN EFFECTS				
UNIT	48.757	3	16.252	1.672 0.171
	48.757	3	16.252	1.672 0.171
EXPLAINED	177.152	5	35.430	3.644 0.003
RESIDUAL	9460.211	973	9.723	
TOTAL	9637.363	978	9.854	

S60140				
SOURCE OF VARIATION				
	SUM OF SQUARES	DF	MEAN SQUARE	SIGNIF. F OF F
COVARIATES				
EAST	237.853	2	118.927	6.617 0.001
NORTH	101.902	1	101.902	5.670 0.017
	128.535	1	128.535	7.151 0.008
MAIN EFFECTS				
UNIT	237.301	3	79.100	4.401 0.004
	237.301	3	79.100	4.401 0.004
EXPLAINED	475.156	5	95.031	5.287 0.000
RESIDUAL	17488.281	973	17.974	
TOTAL	17963.438	978	18.368	



S140230  
SOURCE OF VARIATION

	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
COVARIATES					
EAST	190.571	2	95.285	23.277	0.000
NORTH	62.518	1	62.518	15.272	0.000
	133.600	1	133.600	32.637	0.000
MAIN EFFECTS					
UNIT	67.814	3	22.605	5.522	0.001
	67.814	3	22.605	5.522	0.001
EXPLAINED	258.385	5	51.677	12.624	0.000
RESIDUAL	3983.017	973	4.094		
TOTAL	4241.402	978	4.337		

SAND  
SOURCE OF VARIATION

	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
COVARIATES					
EAST	2508.261	2	1254.130	16.062	0.000
NORTH	828.877	1	828.877	10.615	0.001
	1604.547	1	1604.547	20.549	0.000
MAIN EFFECTS					
UNIT	319.521	3	106.507	1.364	0.252
	319.521	3	106.507	1.364	0.252
EXPLAINED	2827.813	5	565.563	7.243	0.000
RESIDUAL	75974.438	973	78.083		
TOTAL	78802.250	978	80.575		



SILT					
SOURCE OF VARIATION		SUM OF	DF	MEAN	SIGNIF
		SQUARES		SQUARE	OF F
COVARIATES					
EAST		1010.414	2	505.207	7.725 0.000
NORTH		31.047	1	31.047	0.475 0.491
		967.497	1	967.497	14.794 0.000
MAIN EFFECTS					
UNIT		1660.122	3	553.374	8.462 0.000
		1660.121	3	553.374	8.462 0.000
EXPLAINED		2670.535	5	534.107	8.167 0.000
RESIDUAL		63632.527	973	65.398	
TOTAL		66303.063	978	67.795	

CLAY					
SOURCE OF VARIATION		SUM OF	DF	MEAN	SIGNIF
		SQUARES		SQUARE	OF F
COVARIATES					
EAST		632.911	2	316.456	19.917 0.000
NORTH		539.083	1	539.083	33.928 0.000
		60.161	1	60.161	5.945 0.025
MAIN EFFECTS					
UNIT		615.541	3	205.160	12.913 0.000
		615.541	3	205.160	12.913 0.000
EXPLAINED		1248.453	5	249.691	15.715 0.000
RESIDUAL		15459.910	973	15.889	
TOTAL		16708.371	978	17.084	





CAC03		SUE OF		MEAN		SIGNIF	
SOURCE OF VARIATION		SQUARES		SQUARE		OF F	
	DF				P		
COVARIATES							
EAST	2	3161.477		1580.739	353.901	0.000	
NORTH	1	3135.117		3135.117	701.900	0.000	
	1	47.452		47.452	10.624	0.001	
MAIN EFFECTS							
UNIT	3	3588.446		1196.149	267.798	0.000	
	3	3568.849		1196.149	267.798	0.000	
EXPLAINED	5	6749.926		1349.985	302.239	0.0	
RESIDUAL	973	4340.016		4.467			
TOTAL	978	11095.941		11.346			

LOCAL		SUM OF		MEAN		SIGNIF	
SOURCE OF VARIATION		SQUARES		SQUARE		OF F	
	DF				P		
COVARIATES							
EAST	2	5659.645		2829.822	24.439	0.000	
NORTH	1	1144.250		1144.250	9.902	0.002	
	1	4369.512		4369.512	37.613	0.000	
MAIN EFFECTS							
UNIT	3	1996.949		666.316	5.766	0.001	
	3	1998.951		666.317	5.766	0.001	
EXPLAINED	5	7658.625		1531.725	13.255	0.000	
RESIDUAL	973	112435.938		115.556			
TOTAL	978	120094.563		122.796			



ACID					
SOURCE OF VARIATION		SUM OF	MEAN	F	SIGMIF
		SQUARES	SQUARE		OF F
		DF			
COVARIATES					
EAST	2	11779.867	5889.934	40.877	0.000
NORTE	1	11314.367	11314.367	78.523	0.000
	1	332.239	332.239	2.306	0.129
MAIN EFFECTS					
UNIT	3	4747.895	1582.631	10.984	0.000
	3	4747.895	1582.631	10.984	0.000
EXPLAINED	5	16527.813	3305.563	22.941	0.000
RESIDUAL	973	140199.563	144.090		
TOTAL	978	156727.375	160.253		

BASIC					
SOURCE OF VARIATION		SUM OF	MEAN	F	SIGMIF
		SQUARES	SQUARE		OF F
		DF			
COVARIATES					
EAST	2	159.475	79.738	7.719	0.000
NORTH	1	147.072	147.072	14.237	0.000
	1	15.215	15.215	1.473	0.225
MAIN EFFECTS					
UNIT	3	112.527	37.509	3.631	0.013
	3	112.527	37.509	3.631	0.013
EXPLAINED	5	272.004	54.401	5.266	0.000
RESIDUAL	973	10051.227	10.330		
TOTAL	978	10323.230	10.555		



CARB		SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
SOURCE OF VARIATION						
COVARIATES		14.426	2	7.113	0.357	0.700
EAST		0.195	1	0.195	0.007	0.933
NORTH		13.978	1	13.976	0.701	0.403
MAIN EFFECTS						
UNIT		426.560	3	142.187	7.133	0.000
		426.560	3	142.187	7.133	0.000
EXPLAINED		440.789	5	88.158	4.422	0.001
RESIDUAL		19396.598	973	19.935		
TOTAL		19837.387	978	20.284		

QTZ		SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
SOURCE OF VARIATION						
COVARIATES						
EAST		9293.383	2	4646.691	52.505	0.000
NORTH		7233.090	1	7233.090	81.729	0.000
		2307.507	1	2307.507	26.073	0.000
MAIN EFFECTS						
UNIT		4297.668	3	1432.556	16.187	0.000
		4297.664	3	1432.555	16.187	0.000
EXPLAINED		13591.063	5	2718.212	30.714	0.000
RESIDUAL		86110.875	973	88.500		
TOTAL		99701.938	978	101.945		



MM	SOURCE OF VARIATION	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
	COVARIATES					
	EAST	5.802	2	2.901	35.576	0.000
	NORTH	3.976	1	3.976	48.758	0.000
		1.997	1	1.997	24.494	0.000
	MAIN EFFECTS					
	UNIT	0.691	3	0.230	2.824	0.038
		0.691	3	0.230	2.824	0.038
	EXPLAINED	6.492	5	1.298	15.925	0.000
	RESIDUAL	79.335	973	0.082		
	TOTAL	85.828	978	0.088		

MM	SOURCE OF VARIATION	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
	COVARIATES					
	EAST	42.671	2	21.336	160.380	0.000
	NORTH	36.329	1	36.329	307.140	0.000
		7.324	1	7.324	61.918	0.000
	MAIN EFFECTS					
	UNIT	43.062	3	14.354	121.354	0.000
		43.062	3	14.354	121.354	0.000
	EXPLAINED	85.733	5	17.147	144.964	0.000
	RESIDUAL	115.088	973	0.118		
	TOTAL	200.821	978	0.205		





AL.  
SOURCE OF VARIATION

	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
COVARIATES					
EAST	11.628	2	5.814	22.071	0.000
NORTH	11.554	1	11.554	43.876	0.000
	0.144	1	0.144	0.546	0.460
MAIN EFFECTS					
UNIT	9.104	3	3.035	11.524	0.000
	9.104	3	3.035	11.524	0.000
EXPLAINED	20.732	5	4.146	15.746	0.000
RESIDUAL	256.225	973	0.263		
TOTAL	276.957	978	0.283		

SI  
SOURCE OF VARIATION

	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
COVARIATES					
EAST	119.691	2	59.845	60.945	0.000
NORTH	104.605	1	104.605	106.731	0.000
	17.451	1	17.451	17.772	0.000
MAIN EFFECTS					
UNIT	125.256	3	41.752	42.519	0.000
	125.256	3	41.752	42.519	0.000
EXPLAINED	244.947	5	48.989	49.890	0.000
RESIDUAL	955.444	973	0.982		
TOTAL	1200.391	978	1.227		



SOURCE OF VARIATION	N	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
COVARIATES						
EAST	2	0.186	2	0.093	2.851	0.058
NORTH	1	0.153	1	0.153	4.680	0.031
	1	0.038	1	0.038	1.163	0.281
MAIN EFFECTS						
UNIT	3	0.919	3	0.306	9.389	0.000
	3	0.919	3	0.306	9.389	0.000
EXPLAINED	5	1.105	5	0.221	6.774	0.000
RESIDUAL	973	31.741	973	0.033		
TOTAL	978	32.846	978	0.034		

SOURCE OF VARIATION	CN	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF OF F
COVARIATES						
EAST	2	198.232	2	99.116	155.792	0.000
NORTH	1	192.403	1	192.403	302.423	0.000
	1	8.113	1	8.113	12.752	0.000
MAIN EFFECTS						
UNIT	3	141.424	3	47.141	74.098	0.000
	3	141.424	3	47.141	74.098	0.000
EXPLAINED	5	339.656	5	67.931	106.776	0.000
RESIDUAL	973	619.029	973	0.636		
TOTAL	978	958.685	978	0.980		



T1						
SOURCE OF VARIATION	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGMIF OF F	
COVARIATES						
EAST	0.058	2	0.029	7.672	0.000	
NORTH	0.054	1	0.054	14.312	0.000	
	0.005	1	0.005	1.286	0.257	
MAIN EFFECTS						
UNIT	0.223	3	0.074	19.644	0.000	
	0.223	3	0.074	19.644	0.000	
EXPLAINED	0.281	5	0.056	14.855	0.000	
RESIDUAL	3.684	973	0.004			
TOTAL	3.966	978	0.004			

PE	SOURCE OF VARIATION	SUM OF SQUARES	DF	MEAN SQUARE	F	SIGNIF UP P
COVARIATES		6.966	2	3.483	23.753	0.000
	EAST	6.670	1	6.670	45.489	0.000
	NORTH	0.390	1	0.390	2.661	0.103
MAIN EFFECTS		2.566	3	0.855	5.833	0.001
	UNIT	2.566	3	0.855	5.833	0.001
EXPLAINED		9.532	5	1.906	13.001	0.000
RESIDUAL		142.668	973	0.147		
TOTAL		152.199	978	0.156		



## APPENDIX 6

Principal components factor analysis of the 22 analyzed variables, showing the correlation matrix, the pre- and post- rotation factor matrices, the transformation matrix, final communalities for each variable, eigenvalues for each factor and the factscore coefficients.





	S1020	S2040	S4060	S60140	S140230	SAND	SILT	CLAY	LOCAL	ACID	BASIC
S1020	1.00000										
S2040	0.67476	1.00000									
S4060	0.53455	0.30998	1.00000								
S60140	0.34884	0.17486	0.74450	1.00000							
S140230	0.34884	0.17486	0.74450	0.71728	1.00000						
SAND	0.60143	0.34884	0.74450	0.62375	0.83275	1.00000					
SILT	0.46711	0.24556	0.62375	0.46711	0.62375	0.83275	1.00000				
CLAY	0.24556	0.17486	0.62375	0.24556	0.62375	0.83275	0.62375	1.00000			
LOCAL	0.16818	0.34884	0.74450	0.16818	0.74450	0.83275	0.16818	0.34884	1.00000		
ACID	0.16818	0.34884	0.74450	0.16818	0.74450	0.83275	0.16818	0.34884	0.16818	1.00000	
BASIC	0.16818	0.34884	0.74450	0.16818	0.74450	0.83275	0.16818	0.34884	0.16818	0.16818	1.00000
CARR	0.10547	0.07476	0.05345	0.03484	0.02375	0.01682	0.01055	0.00627	0.00397	0.00254	0.00168
QTZ	0.07476	0.05345	0.03484	0.02375	0.01682	0.01055	0.00627	0.00397	0.00254	0.00168	0.00105
CAO3	0.05345	0.03484	0.02375	0.01682	0.01055	0.00627	0.00397	0.00254	0.00168	0.00105	0.00063
W	0.03484	0.02375	0.01682	0.01055	0.00627	0.00397	0.00254	0.00168	0.00105	0.00063	0.00039
AL	0.02375	0.01682	0.01055	0.00627	0.00397	0.00254	0.00168	0.00105	0.00063	0.00039	0.00025
SI	0.01682	0.01055	0.00627	0.00397	0.00254	0.00168	0.00105	0.00063	0.00039	0.00025	0.00016
K	0.01055	0.00627	0.00397	0.00254	0.00168	0.00105	0.00063	0.00039	0.00025	0.00016	0.00010
CA	0.00627	0.00397	0.00254	0.00168	0.00105	0.00063	0.00039	0.00025	0.00016	0.00010	0.00006
TI	0.00397	0.00254	0.00168	0.00105	0.00063	0.00039	0.00025	0.00016	0.00010	0.00006	0.00003
FE	0.00254	0.00168	0.00105	0.00063	0.00039	0.00025	0.00016	0.00010	0.00006	0.00003	0.00001



## FACTOR MATRIX USING PRINCIPAL FACTOR, NO ITERATIONS

	FACTOR 1	FACTOR 2	FACTOR 3	FACTOR 4	FACTOR 5	FACTOR 6	FACTOR 7
S1020	0.69394	-0.04289	0.05684	0.01380	0.28143	0.35729	-0.17624
S2040	0.71882	-0.32287	0.10378	0.04752	0.15167	0.23618	-0.22591
S4060	0.59233	-0.38827	0.04951	0.06383	-0.03504	0.32747	-0.18607
S60140	0.68020	-0.34860	0.16909	-0.12240	-0.13595	-0.14217	0.27548
S140230	0.29273	-0.37498	0.00034	0.12852	0.07555	-0.37202	0.60489
SAND	0.86058	-0.46009	0.12699	0.00390	0.01872	0.06777	0.13693
SILT	-0.74693	0.45492	-0.12302	-0.28364	0.05025	-0.10651	-0.18991
CLAY	-0.36895	0.07938	-0.01617	0.58553	-0.14449	0.07379	0.09276
LOCAL	-0.45458	0.04057	-0.23220	-0.34902	0.24803	0.57045	0.45544
ACID	0.39150	0.24389	-0.28528	0.66497	-0.27678	-0.11150	-0.20850
BASIC	0.17940	0.09679	-0.05704	0.14441	0.17365	-0.30540	-0.04741
CARB	0.04739	0.02924	0.32374	-0.22180	0.46348	-0.19785	-0.33587
QTZ	-0.04650	-0.40122	0.51245	-0.39308	-0.19698	-0.33570	-0.09893
CAC03	0.53169	0.65231	-0.11182	-0.02342	0.16935	-0.09204	-0.03265
NA	0.26116	0.35231	0.31206	-0.39724	-0.35973	-0.03316	0.05524
MG	0.45676	0.80132	0.01947	-0.00820	0.10513	-0.04452	0.14314
AL	-0.46308	0.02431	0.54349	0.35201	-0.04796	0.21083	0.09604
SI	-0.19755	-0.76249	-0.50843	-0.13589	-0.08282	-0.05400	-0.17708
K	0.13001	0.38564	0.39176	-0.13764	-0.51275	-0.27425	-0.10759
CA	0.53397	0.73244	0.02192	-0.01887	0.19556	-0.08186	0.13175
TI	-0.33534	-0.25374	0.32672	0.18307	0.36458	-0.10448	-0.10178
FE	-0.41284	-0.04551	0.61778	0.34371	0.20453	0.06605	0.09959

## VARIMAX ROTATED FACTOR MATRIX

	FACTOR 1	FACTOR 2	FACTOR 3	FACTOR 4	FACTOR 5	FACTOR 6	FACTOR 7
S1020	0.76999	0.30475	-0.10018	-0.05529	0.10366	-0.11861	-0.03243
S2040	0.84844	0.04223	-0.10128	-0.02913	0.12729	0.00899	0.12156
S4060	0.78278	-0.11767	-0.10106	0.05508	-0.04104	0.00196	0.05148
S60140	0.53560	0.00993	-0.18474	0.21273	0.09040	0.59762	0.10330
S140230	0.13377	-0.02247	-0.00372	-0.20554	-0.05268	0.82931	0.04040
SAND	0.85964	0.01856	-0.16366	0.05770	0.06066	0.45828	0.09559
SILT	-0.80779	0.01626	-0.01699	0.00684	0.17352	-0.44746	-0.16293
CLAY	-0.23589	-0.07442	0.41323	-0.14042	-0.49590	-0.08835	0.12253
LOCAL	-0.21722	-0.04324	-0.00443	-0.13690	-0.10909	-0.09517	-0.94318
ACID	0.18490	0.27260	-0.09761	-0.09270	-0.57576	-0.10961	0.63142
BASIC	-0.00367	0.21887	-0.05282	-0.20378	0.08403	0.10745	0.28371
CARB	0.05331	0.13539	0.12175	-0.11018	0.66491	-0.14125	0.11110
QTZ	-0.00719	-0.42486	0.09816	0.39166	0.53633	0.27999	0.15539
CAC03	0.10349	0.80009	-0.27644	0.03990	0.02490	-0.07053	0.15512
NA	-0.01526	0.27727	-0.13992	0.67493	0.14969	0.02015	-0.00739
MG	-0.00399	0.91111	-0.13326	0.17638	-0.04819	-0.00580	0.05798
AL	-0.15158	-0.11713	0.80582	0.16226	-0.09722	-0.08579	-0.07976
SI	0.05561	-0.80523	-0.40577	-0.34797	-0.01847	0.02231	-0.02151
K	0.06226	0.18410	0.10001	0.76125	-0.10658	-0.22119	0.01338
CA	0.07637	0.91444	-0.15293	0.10430	0.02885	0.04071	0.07734
TI	-0.06578	-0.21349	0.49445	-0.29166	0.29900	-0.01770	0.04610
FE	-0.11695	-0.06494	0.83482	-0.03166	0.10319	0.01163	-0.04205

## TRANSFORMATION MATRIX

	FACTOR 1	FACTOR 2	FACTOR 3	FACTOR 4	FACTOR 5	FACTOR 6	FACTOR 7
FACTOR 1	0.74967	0.41511	-0.34534	0.12922	0.01448	0.26755	0.24509
FACTOR 2	-0.39821	0.82018	-0.01804	0.23511	-0.11363	-0.31580	0.02206
FACTOR 3	0.14861	0.05334	0.73318	0.48754	0.42272	0.12527	0.07377
FACTOR 4	0.12508	0.07881	0.51848	-0.37405	-0.59797	-0.01384	0.46005
FACTOR 5	0.12242	0.32537	0.18684	-0.72974	0.49409	-0.05613	-0.25256
FACTOR 6	0.45092	-0.02956	0.16101	0.13467	-0.33591	-0.47816	-0.64051
FACTOR 7	-0.15310	0.19420	0.11490	-0.00292	-0.30603	0.76398	-0.49831



VARIABLE	MEAN	STANDARD DEV	CASES	COMMUNALITY	FACTOR	EIGENVALUE	PCT OF VAR	CUM PCT
SI020	3.1363	1.5109	995	0.72473	1	5.12665	23.3	23.3
SI040	5.3427	1.9480	995	0.76380	2	3.73185	16.8	40.1
SI050	8.7190	3.1687	995	0.64418	3	1.97412	8.8	48.9
SI06140	14.7073	4.3612	995	0.74235	4	1.71381	7.8	56.7
SI10230	5.7508	2.1169	995	0.75282	5	1.22749	5.6	62.3
SAND	37.6582	9.1969	995	0.92228	6	1.20968	5.5	67.8
SILT	51.4538	6.5587	995	0.91000	7	1.13463	5.0	72.8
CLAY	10.8831	4.1371	995	0.52039	8	0.95712	4.4	77.2
LOCAL	7.3063	11.7976	995	0.97837	9	0.93713	4.3	81.5
ACID	59.0762	12.8700	995	0.85983	10	0.78501	3.6	85.2
BASIC	4.2544	3.2475	995	0.91133	11	0.65858	3.0	88.1
CARB	4.0168	4.4792	995	0.52371	12	0.59354	2.7	90.8
QTZ	25.3226	10.1339	995	0.73378	13	0.49000	2.2	93.0
CAC03	3.3571	3.3571	995	0.75947	14	0.47255	2.1	95.1
NA	0.3125	0.2949	995	0.58106	15	0.38936	1.8	96.9
MG	1.0519	0.4511	995	0.88472	16	0.28885	1.3	98.2
AL	7.6266	0.5492	995	0.73539	17	0.25396	1.2	99.4
SI	33.1073	1.1366	995	0.93852	18	0.09541	0.4	99.8
K	2.0373	0.2031	995	0.68774	19	0.04945	0.2	100.0
CA	2.9059	0.9876	995	0.88474	20	0.00257	0.0	100.0
TI	0.4481	0.0642	995	0.87130	21	0.00000	0.0	100.0
FE	3.8204	0.4002	995	0.72839	22	-0.00279	-0.0	100.0

## FACTOR SCORE COEFFICIENTS

	FACTOR 1	FACTOR 2	FACTOR 3	FACTOR 4	FACTOR 5	FACTOR 6	FACTOR 7
SI020	0.29714	0.08414	0.05125	-0.10112	0.07377	-0.23322	-0.12879
SI040	0.28576	-0.01331	0.03789	-0.05002	0.07610	-0.05002	-0.07520
SI050	0.27844	-0.04017	0.02105	0.04543	-0.05689	0.17854	-0.07820
SI06140	0.03628	-0.00724	-0.02801	0.12847	0.00072	0.32986	-0.071599
SI10230	-0.12243	0.08212	0.04693	-0.13239	0.00630	0.00792	-0.04316
SAND	0.18554	-0.00101	0.01970	-0.01952	0.00627	0.17359	-0.05721
SILT	-0.19632	0.00649	-0.11053	-0.00913	0.15414	-0.01745	0.01745
CLAY	-0.02074	-0.00978	0.10295	-0.04218	-0.31567	0.01016	0.00421
LOCAL	0.06011	-0.08171	-0.00213	-0.07630	-0.11625	0.03919	-0.69258
ACID	0.01735	0.00131	-0.01319	-0.03908	-0.32344	0.11208	0.70784
BASIC	-0.06797	0.08637	-0.00964	0.17237	0.10252	0.07605	0.10311
CARB	0.03159	0.07808	0.06165	-0.16365	0.48182	0.15267	0.11511
QTZ	-0.08329	-0.00151	0.26613	0.26613	0.29919	0.14142	0.17501
NA	-0.01563	0.22507	-0.07834	0.07919	0.06798	-0.02942	0.04737
CAC03	-0.06062	0.00448	-0.07462	0.40430	0.04583	0.07450	-0.07149
MG	-0.04450	0.26899	-0.00882	0.00117	-0.00114	0.06818	-0.03742
AL	0.06070	-0.00085	0.38852	0.01195	-0.00114	-0.00556	-0.06334
SI	0.06631	-0.25708	-0.25504	-0.10181	-0.10181	-0.07423	0.05566
K	0.06358	-0.06147	0.04350	0.49220	-0.13058	-0.15910	-0.07049
CA	-0.02968	0.28233	-0.00436	-0.05020	0.05535	0.08096	-0.07067
TI	0.07425	0.01558	0.23407	-0.21021	0.21874	-0.02181	0.07028
FE	0.04810	0.05952	0.41644	-0.04701	0.05177	0.05390	-0.07616

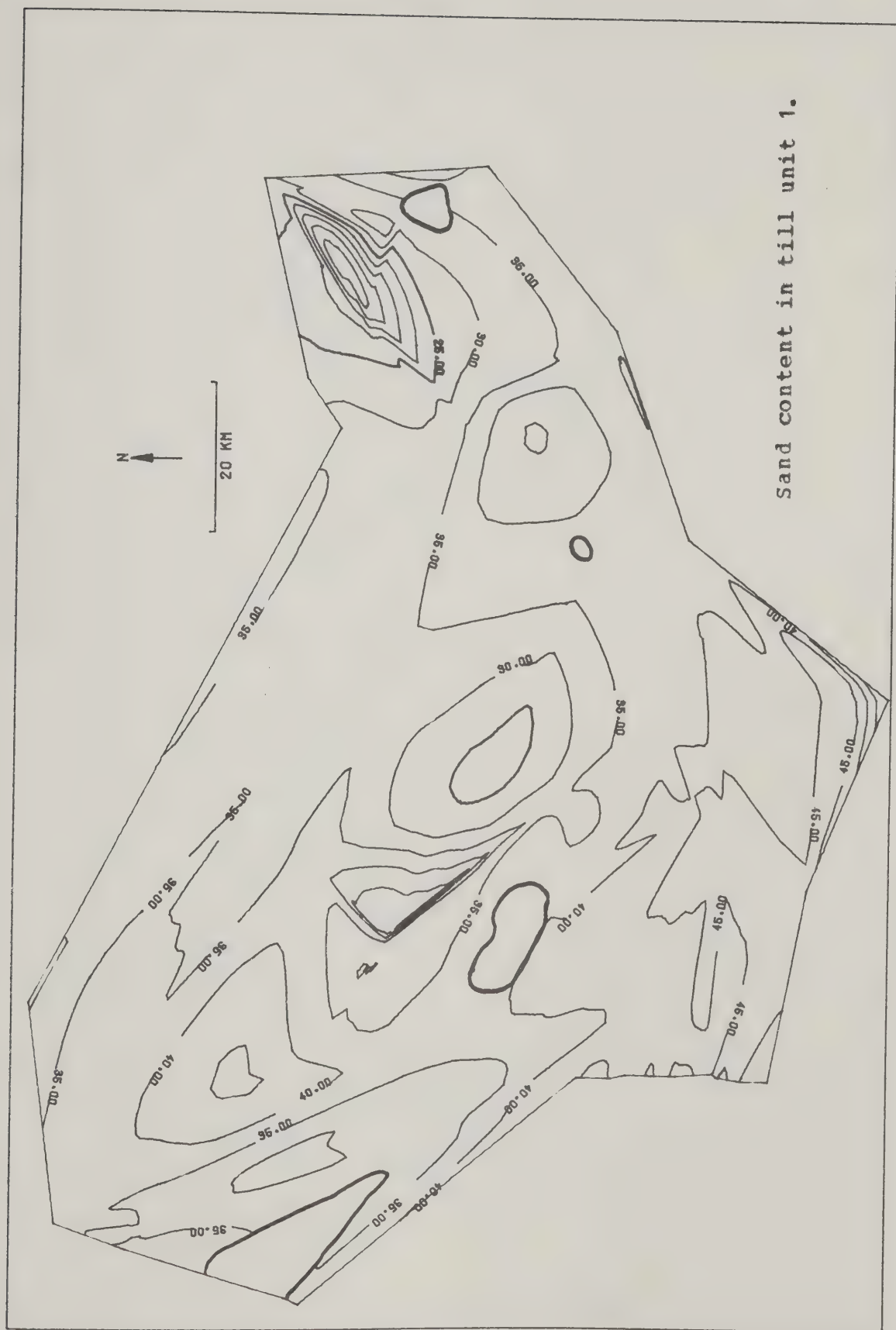


## APPENDIX 7

Contour maps showing the lateral variation of sand, clay, grain lithology, calcium carbonate equivalent and bulk chemistry in each of the four till units. A heavy black line marks any till unit boundaries.

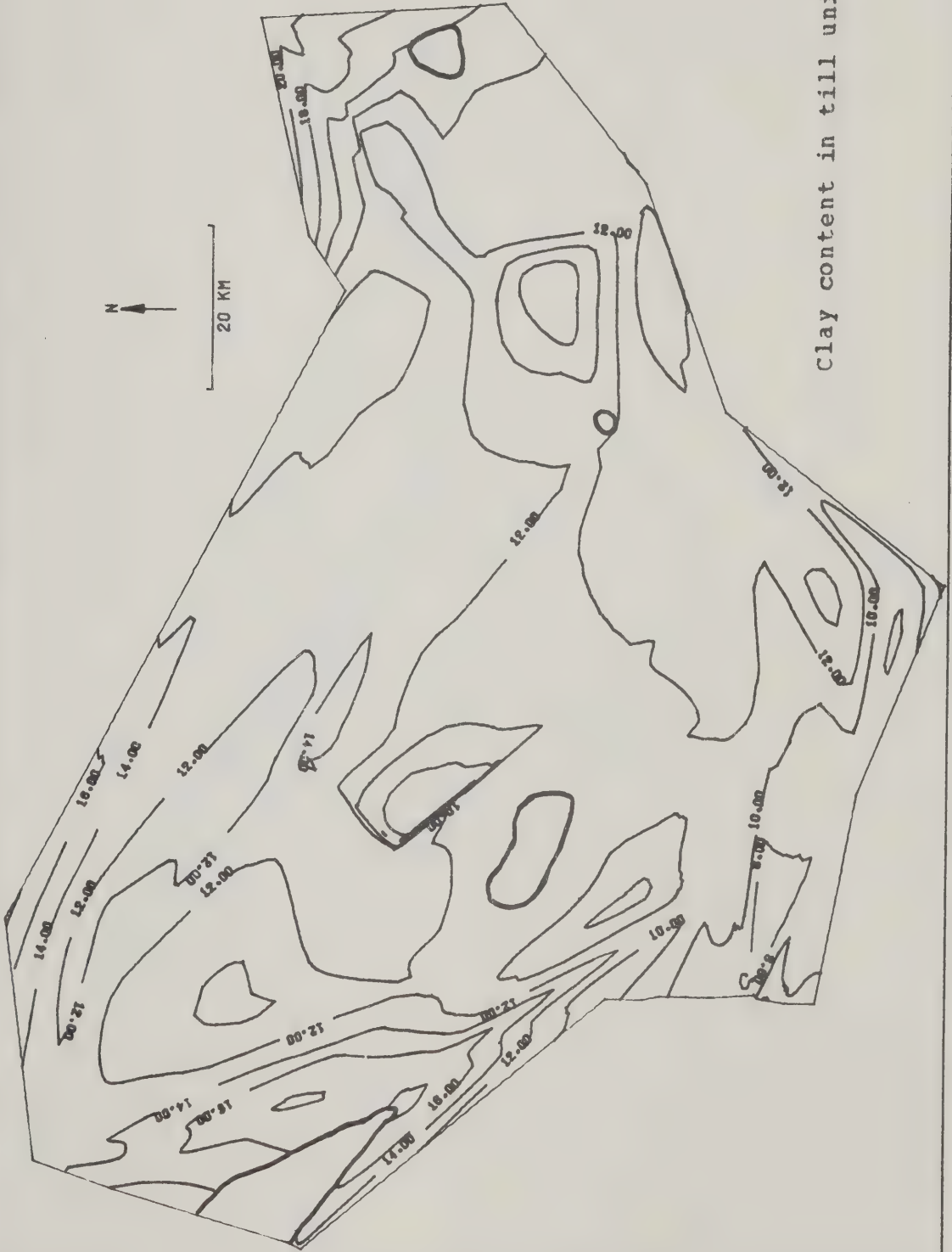






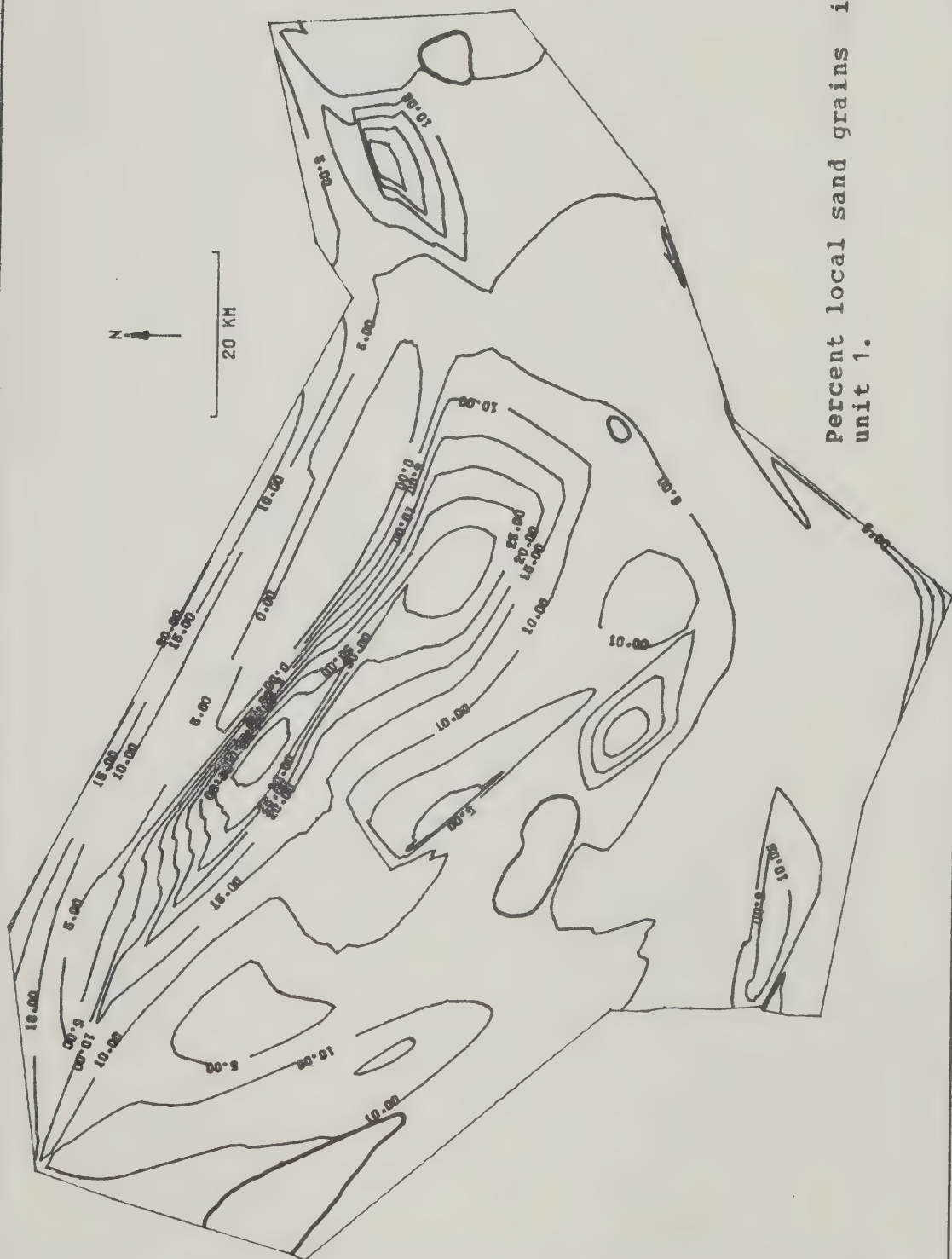


Clay content in till unit 1.

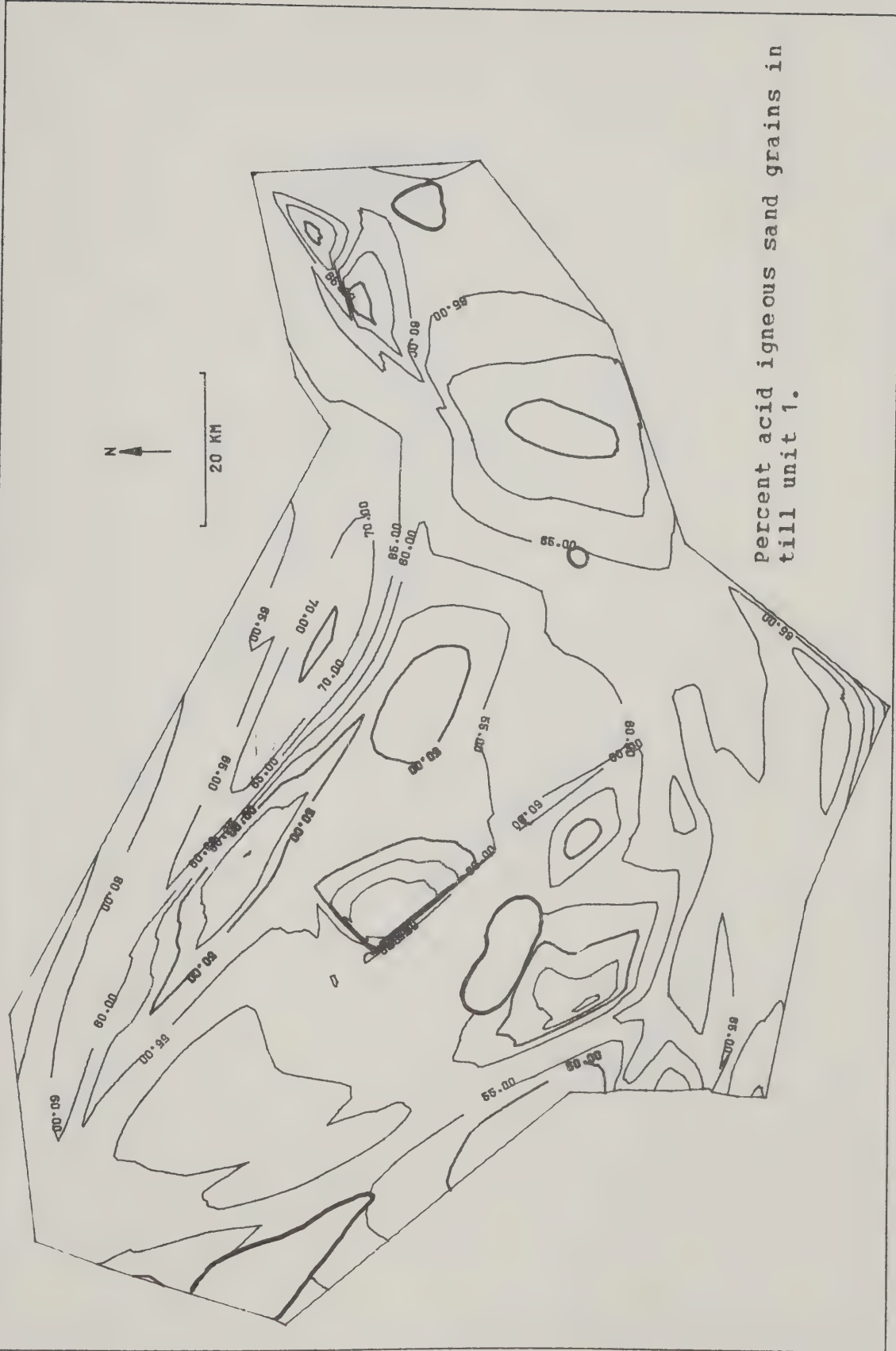




Percent local sand grains in till  
unit 1.





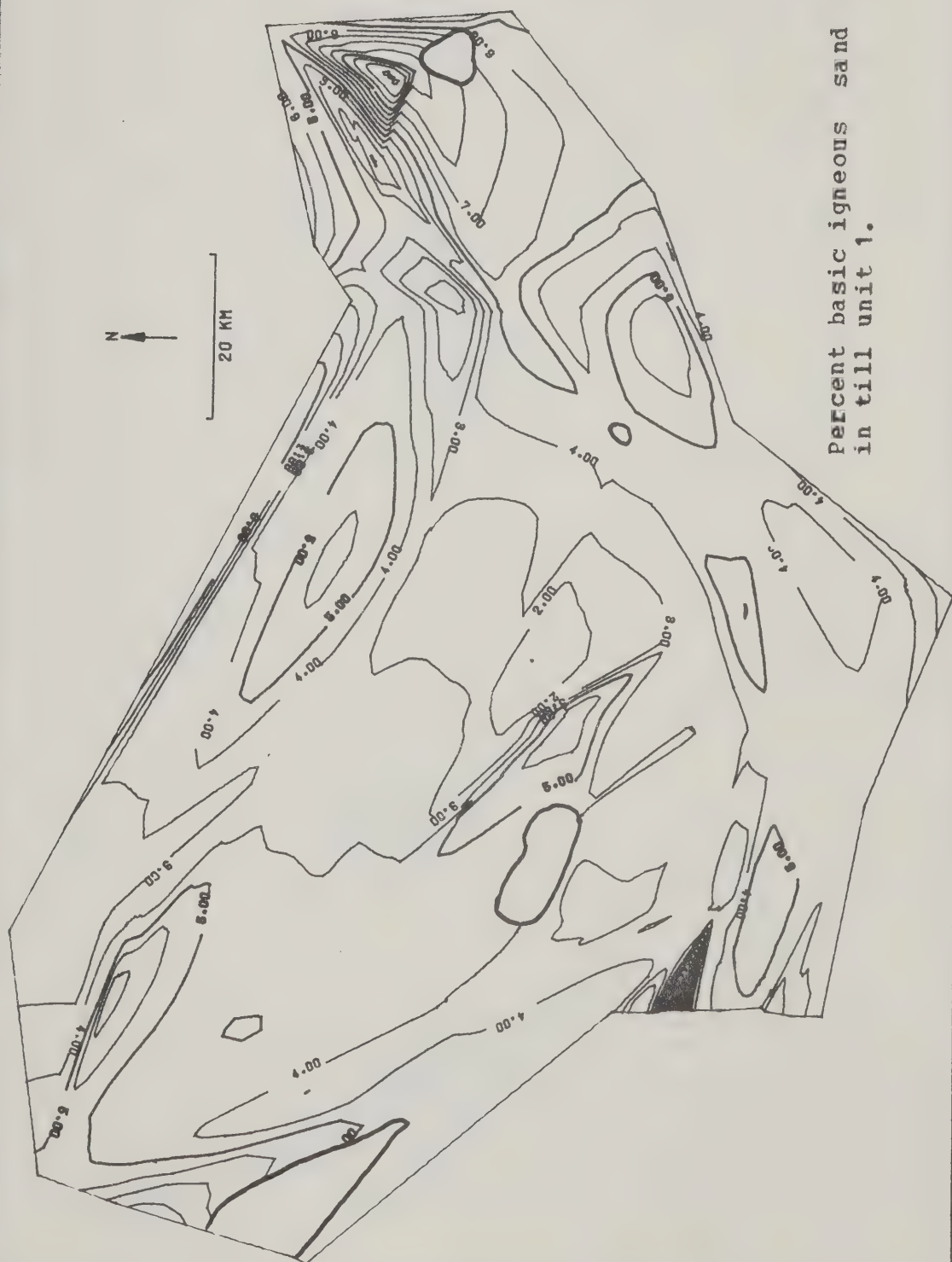


Percent acid igneous sand grains in till unit 1.



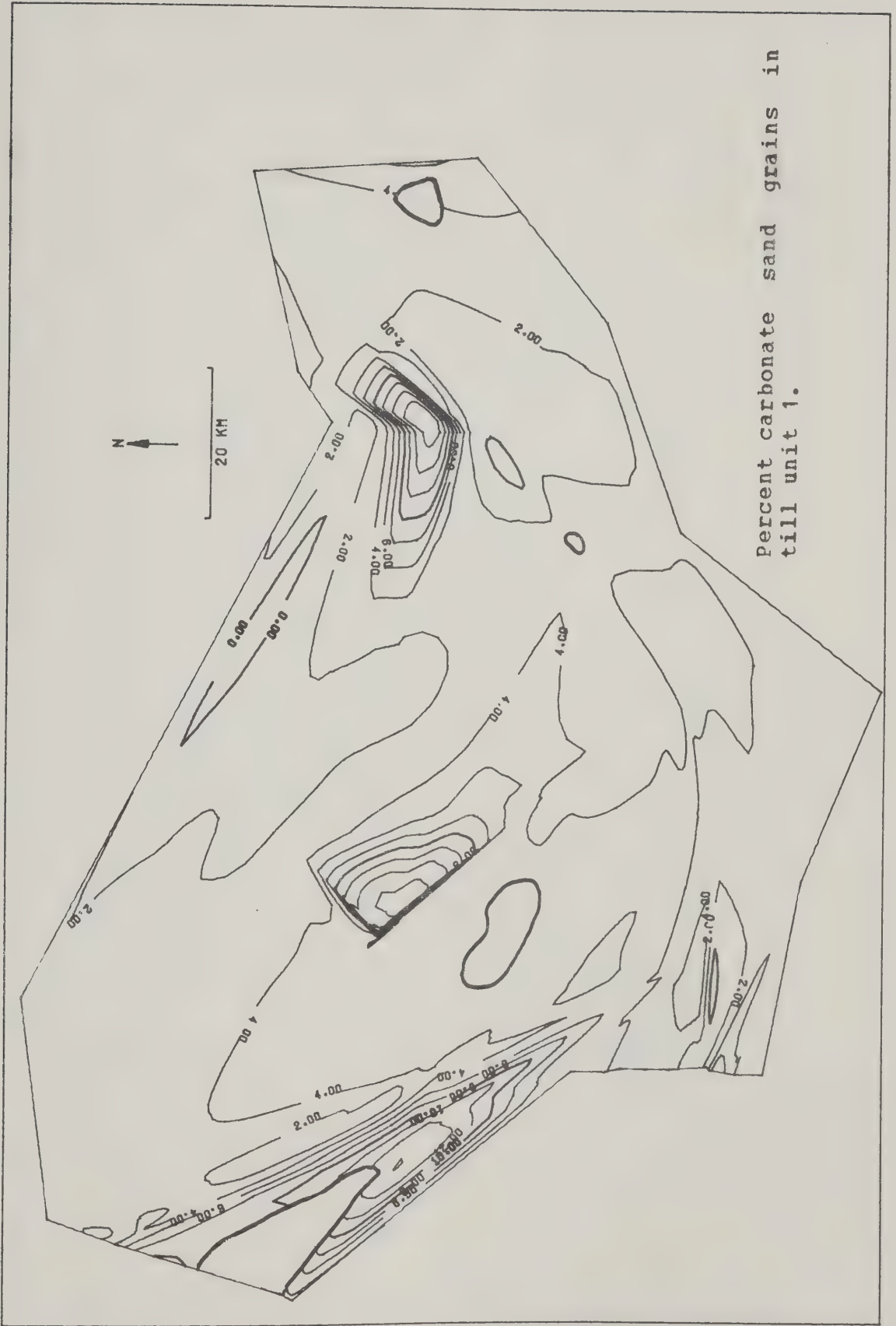


Percent basic igneous sand grains  
in till unit 1.

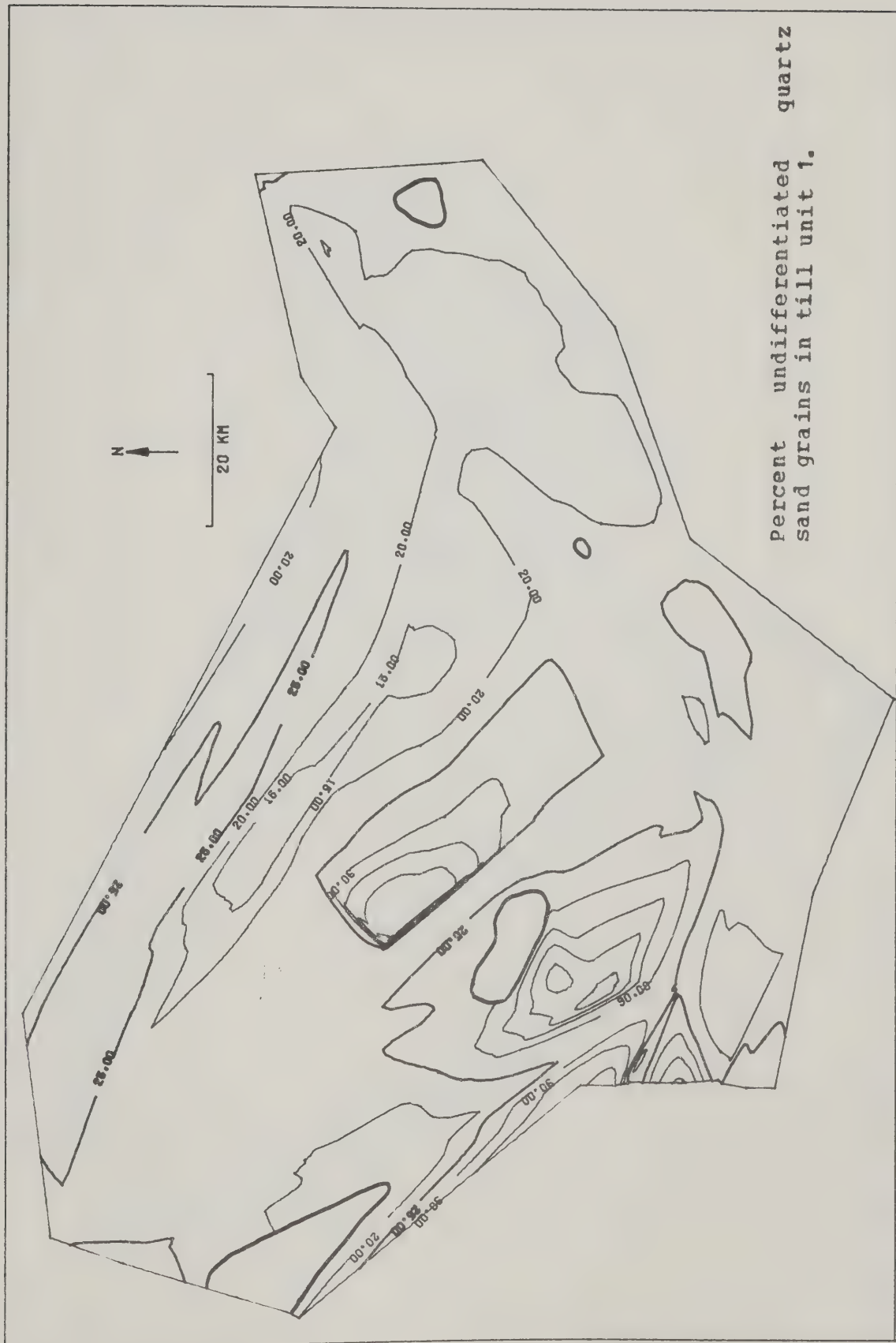




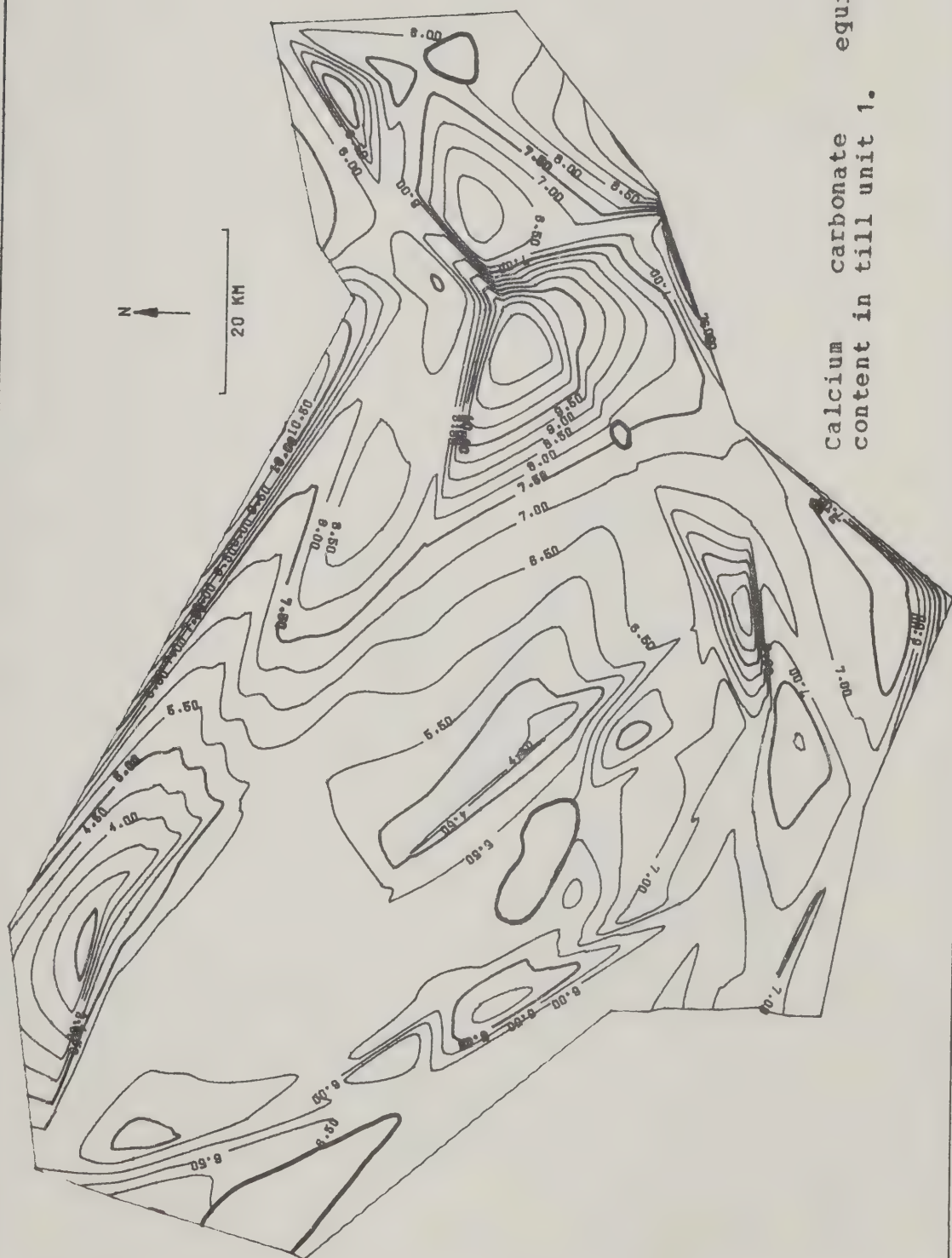
Percent carbonate sand grains in  
till unit 1.







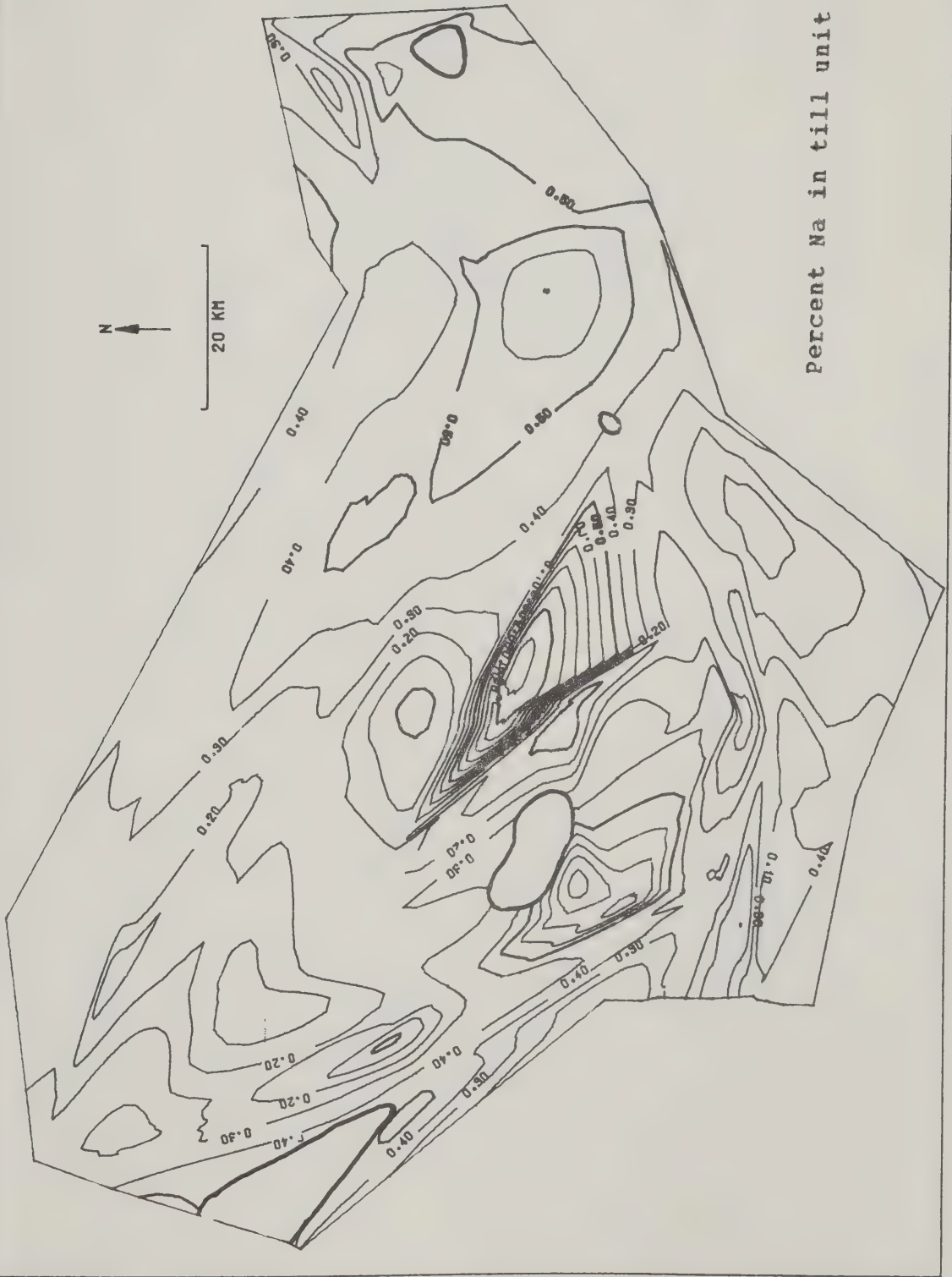




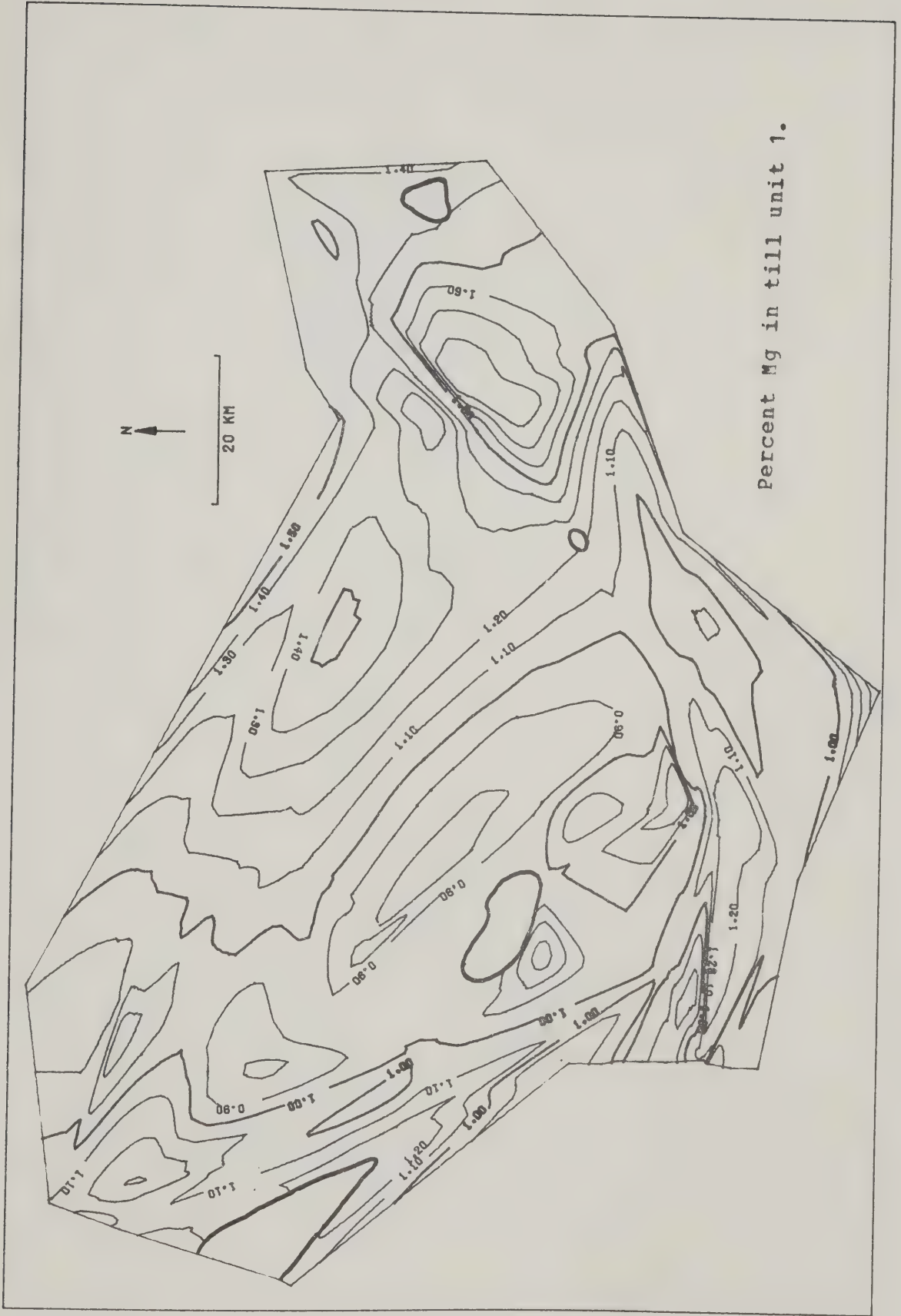




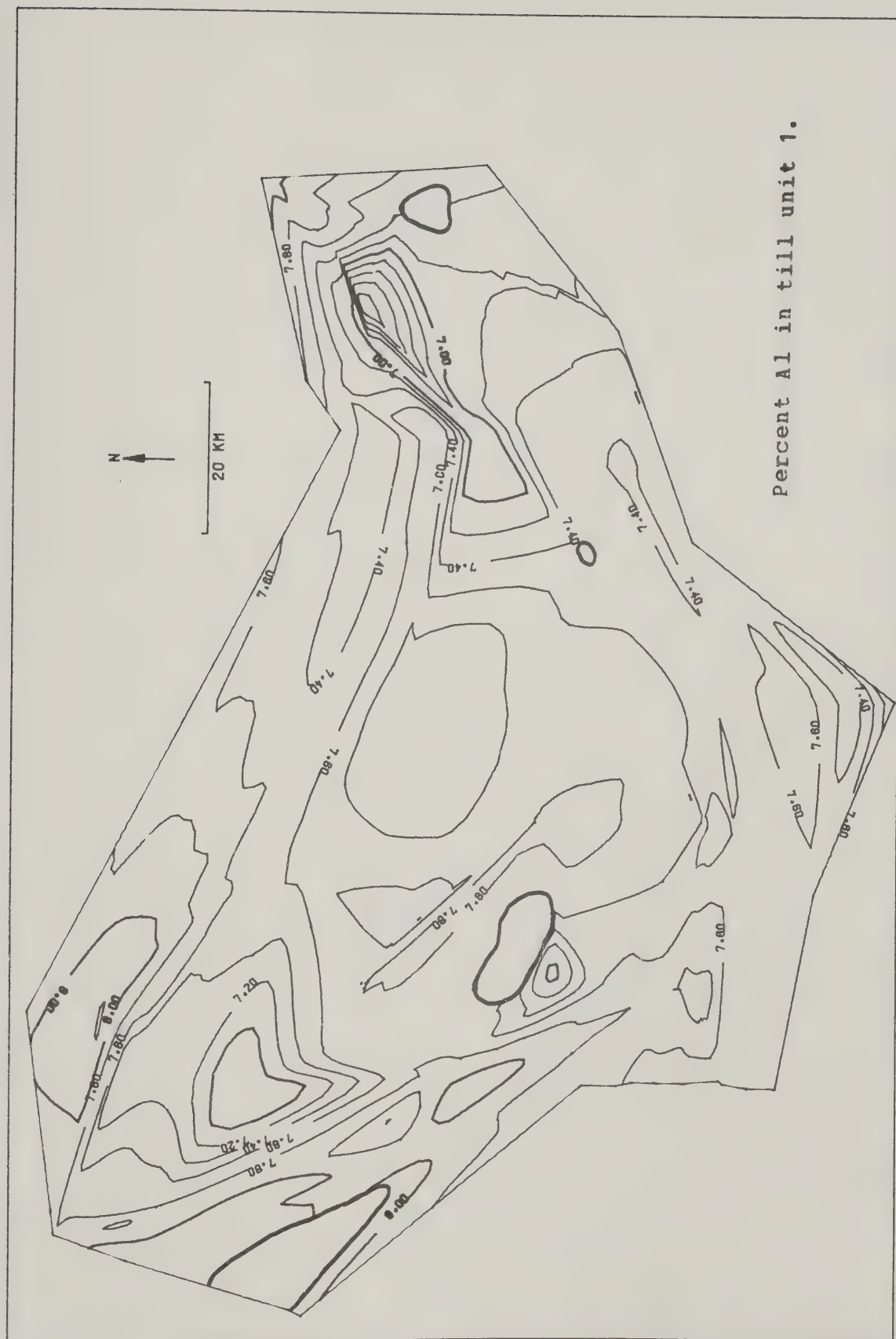
Percent Na in till unit 1.



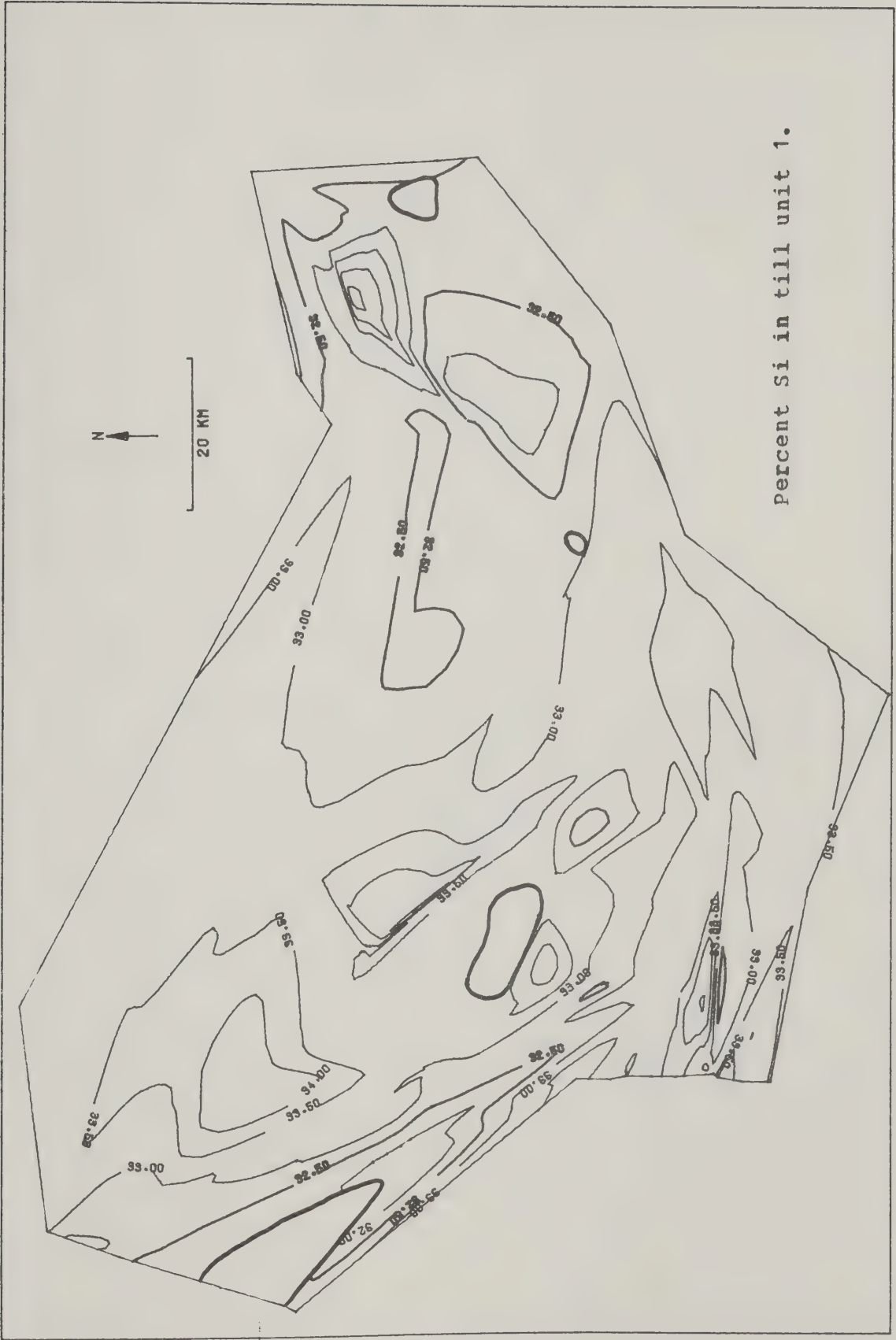








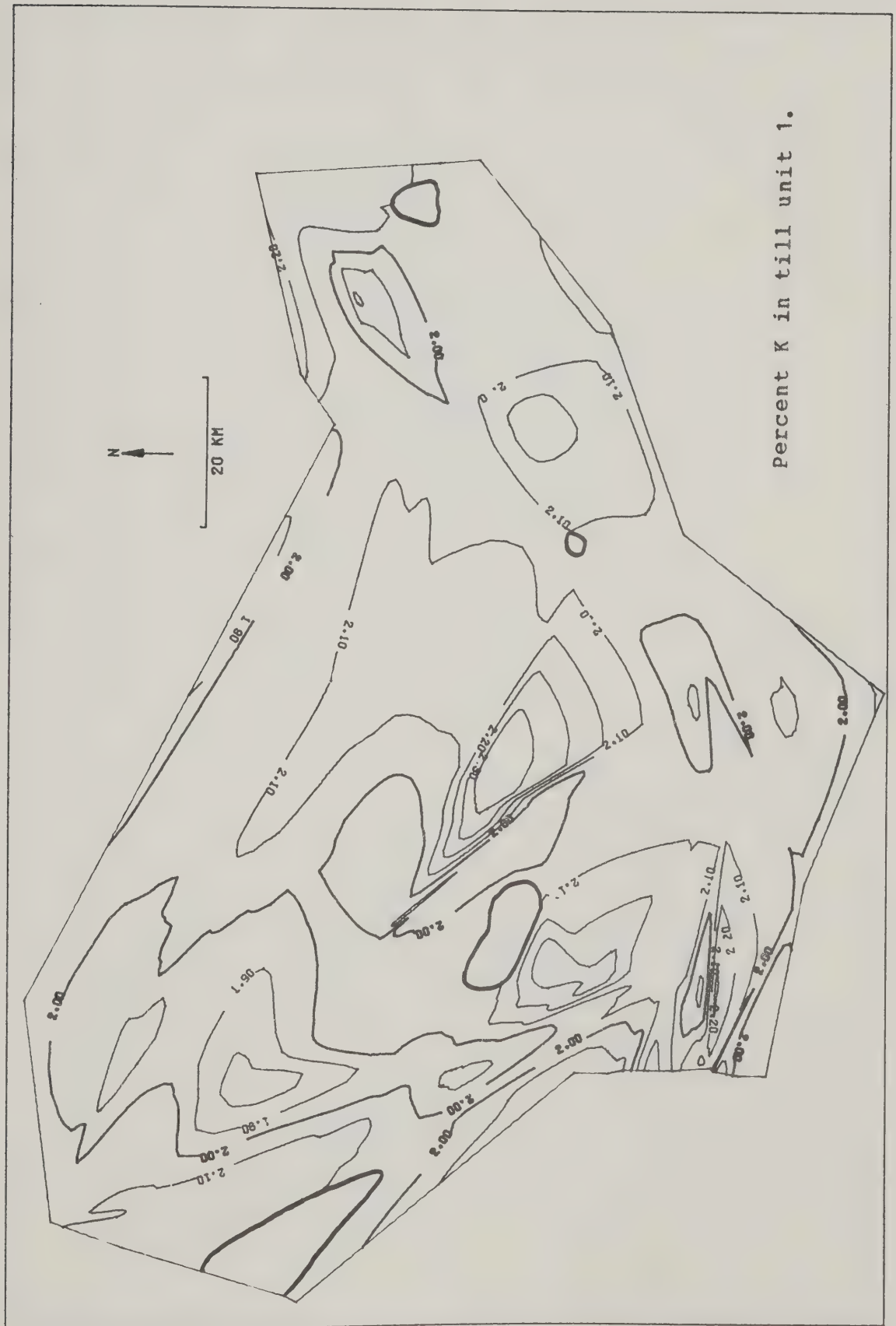




Percent Si in till unit 1.





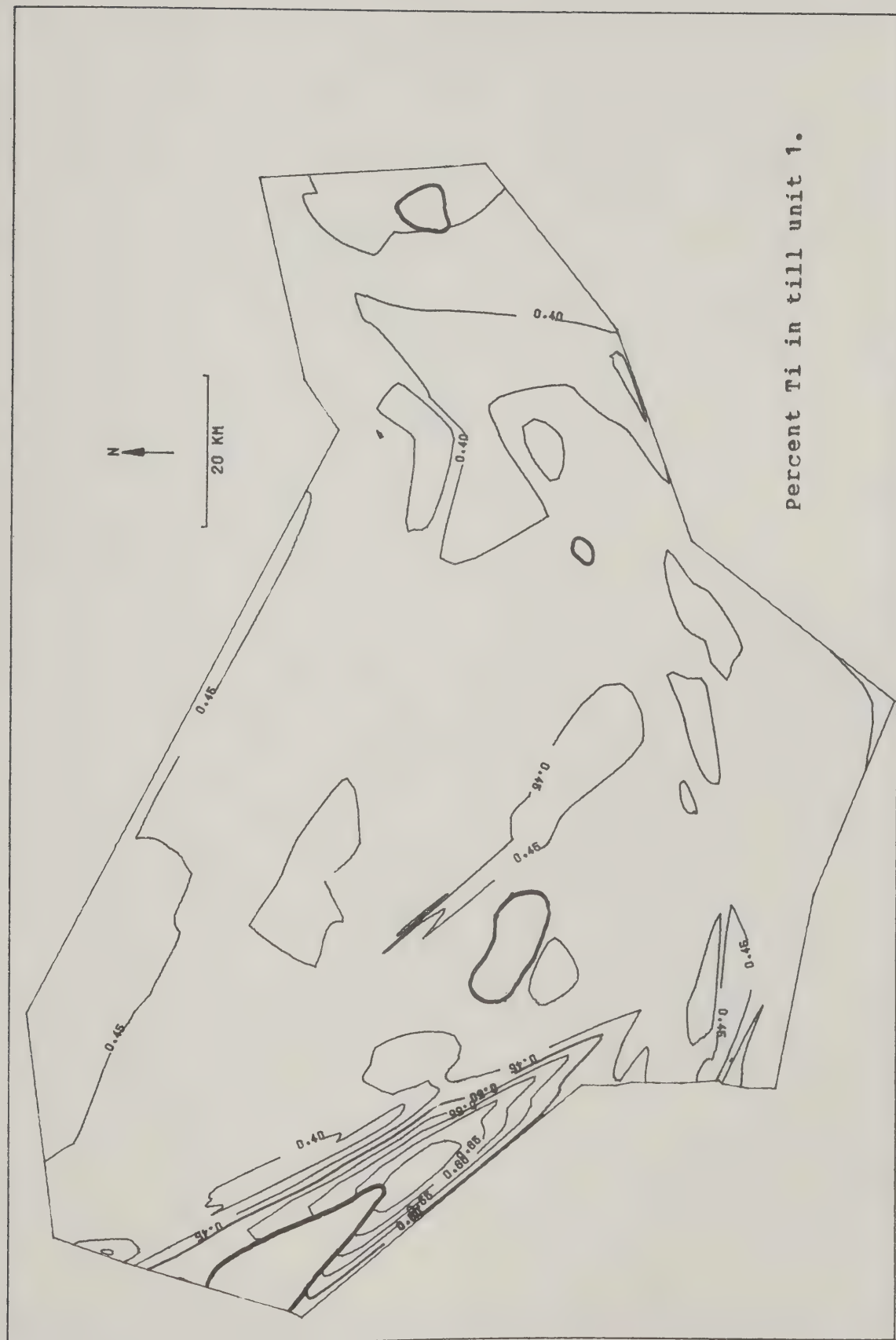




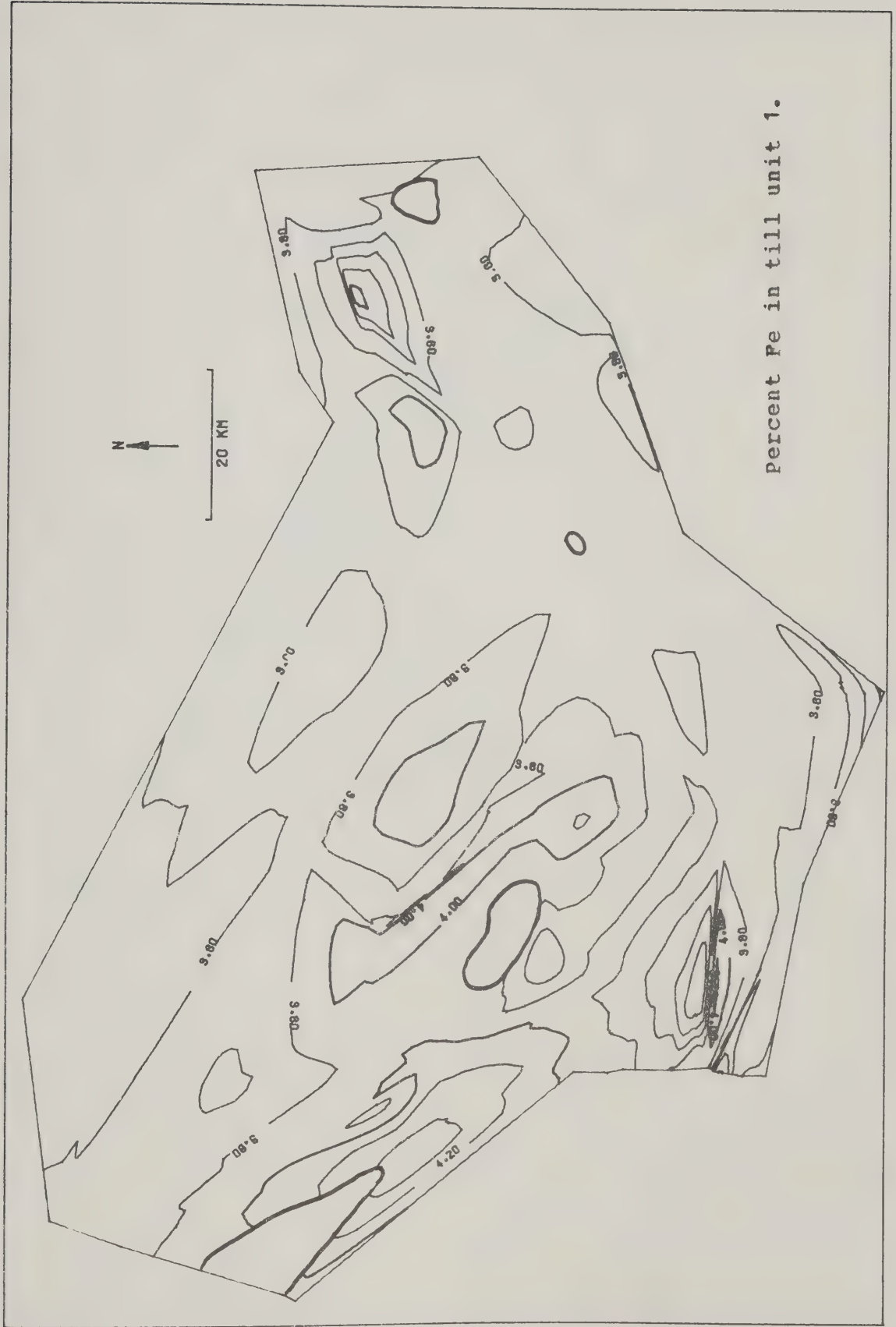
Percent Ca in till unit 1.











Percent Fe in till unit 1.



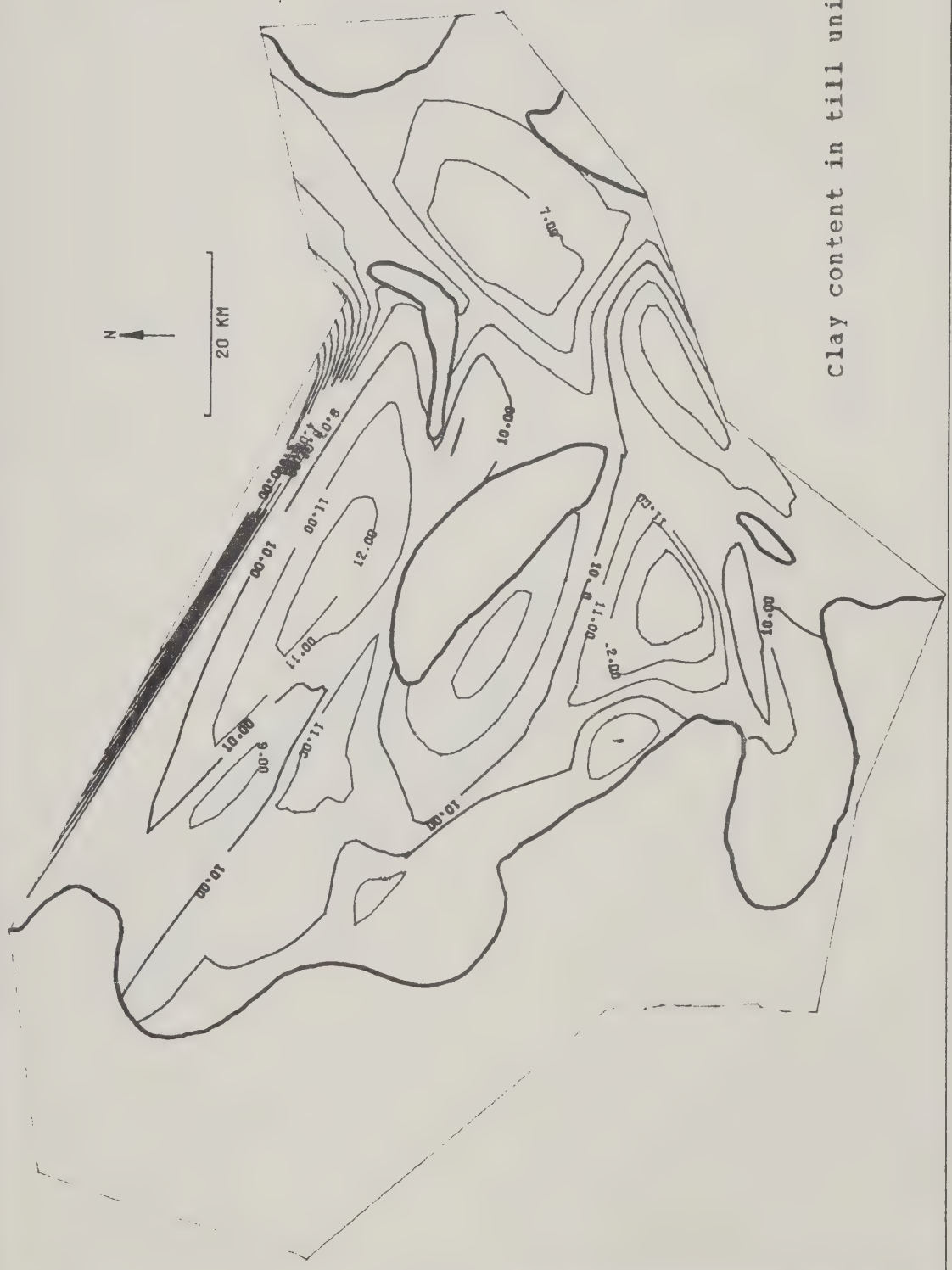


Sand content in till unit 2.

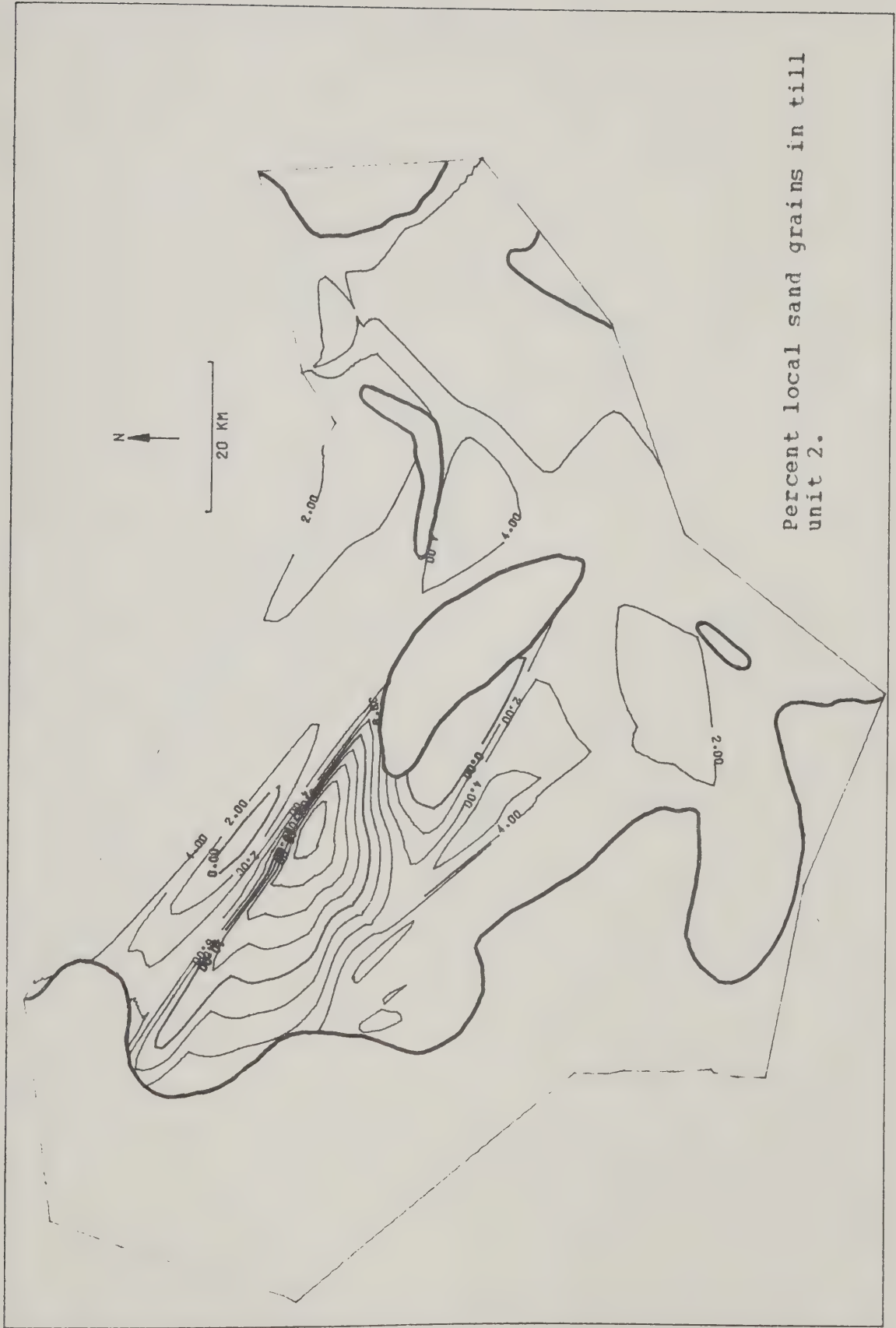




Clay content in till unit 2.

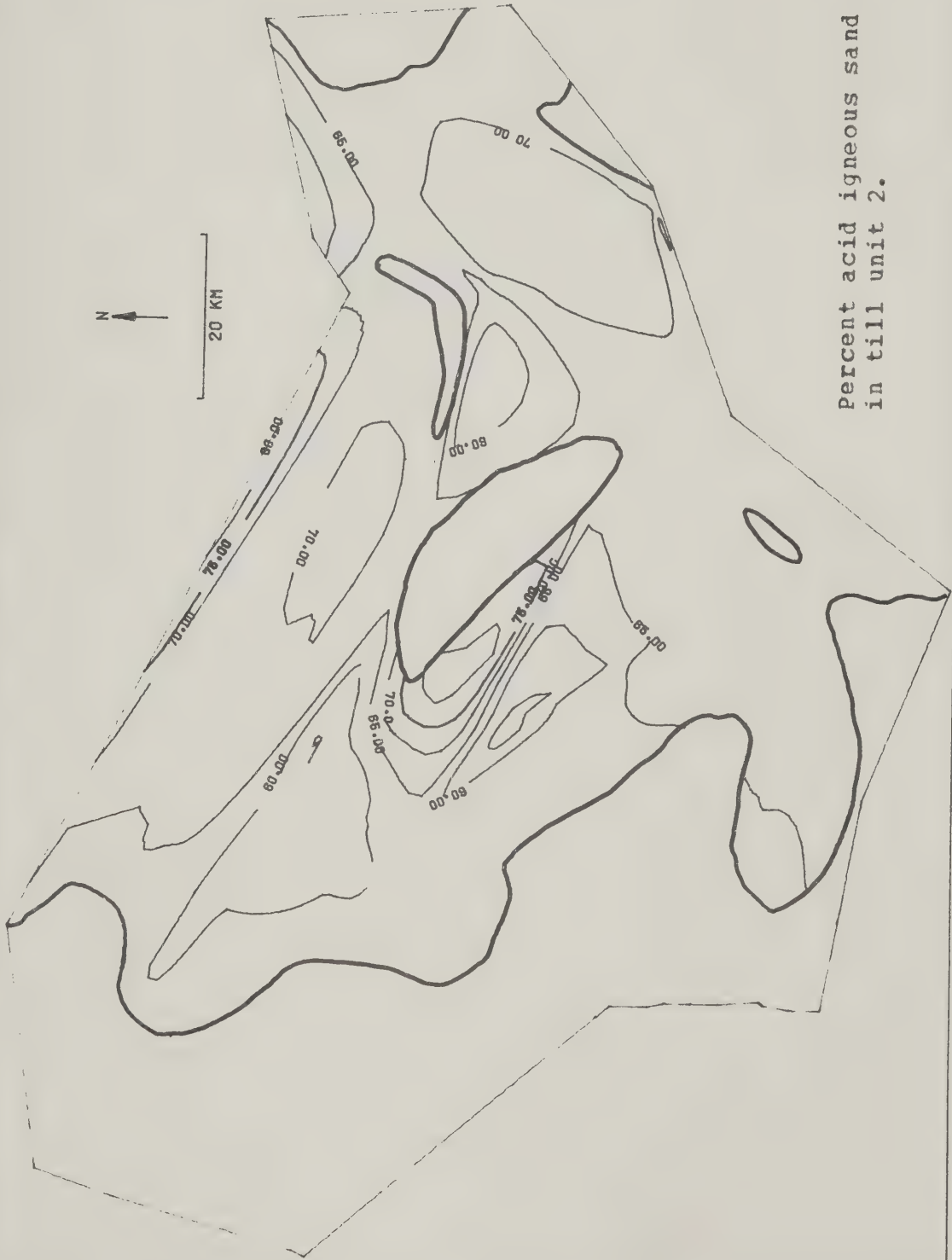






Percent local sand grains in till  
unit 2.



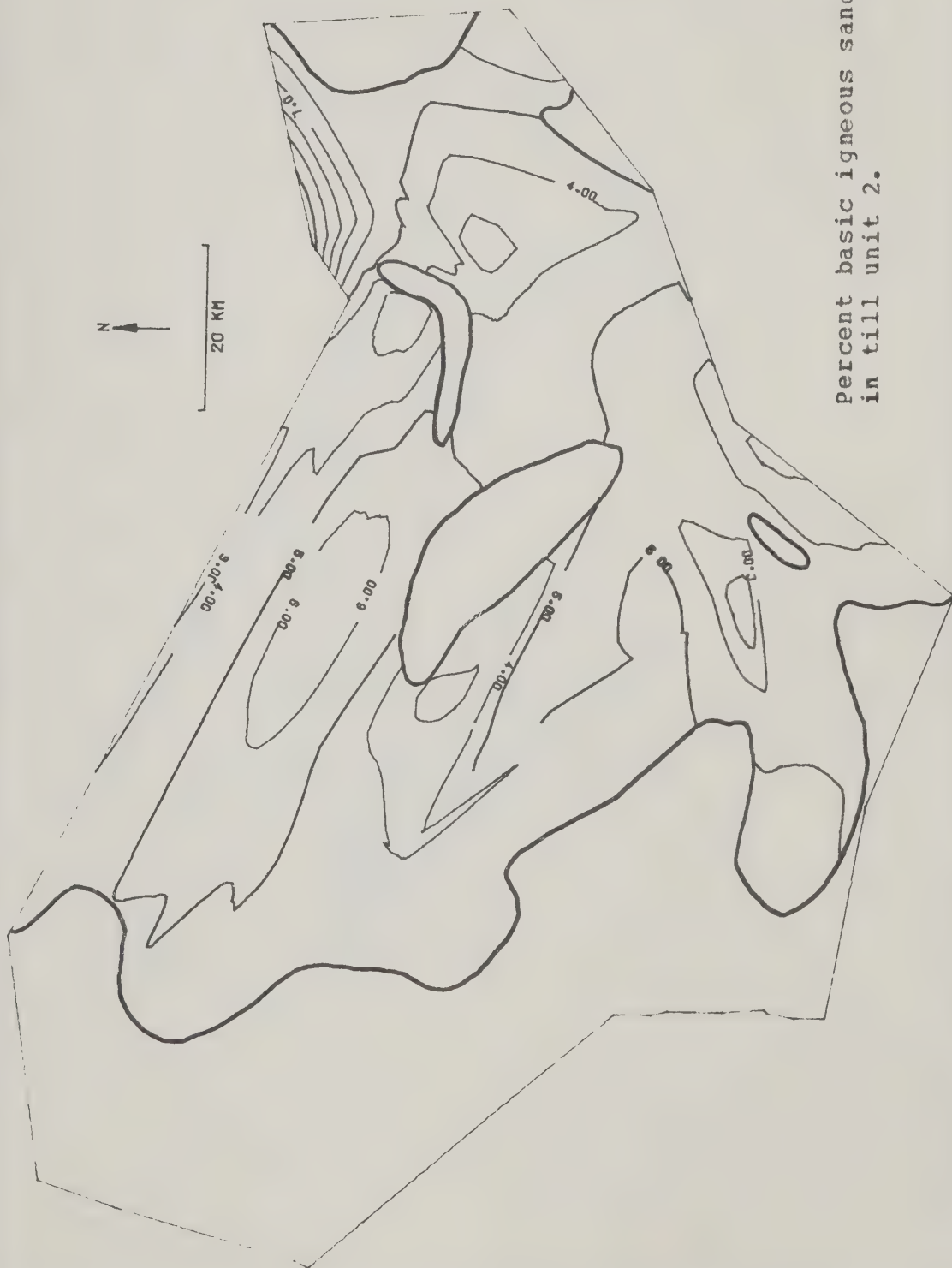


Percent acid igneous sand grains  
in till unit 2.



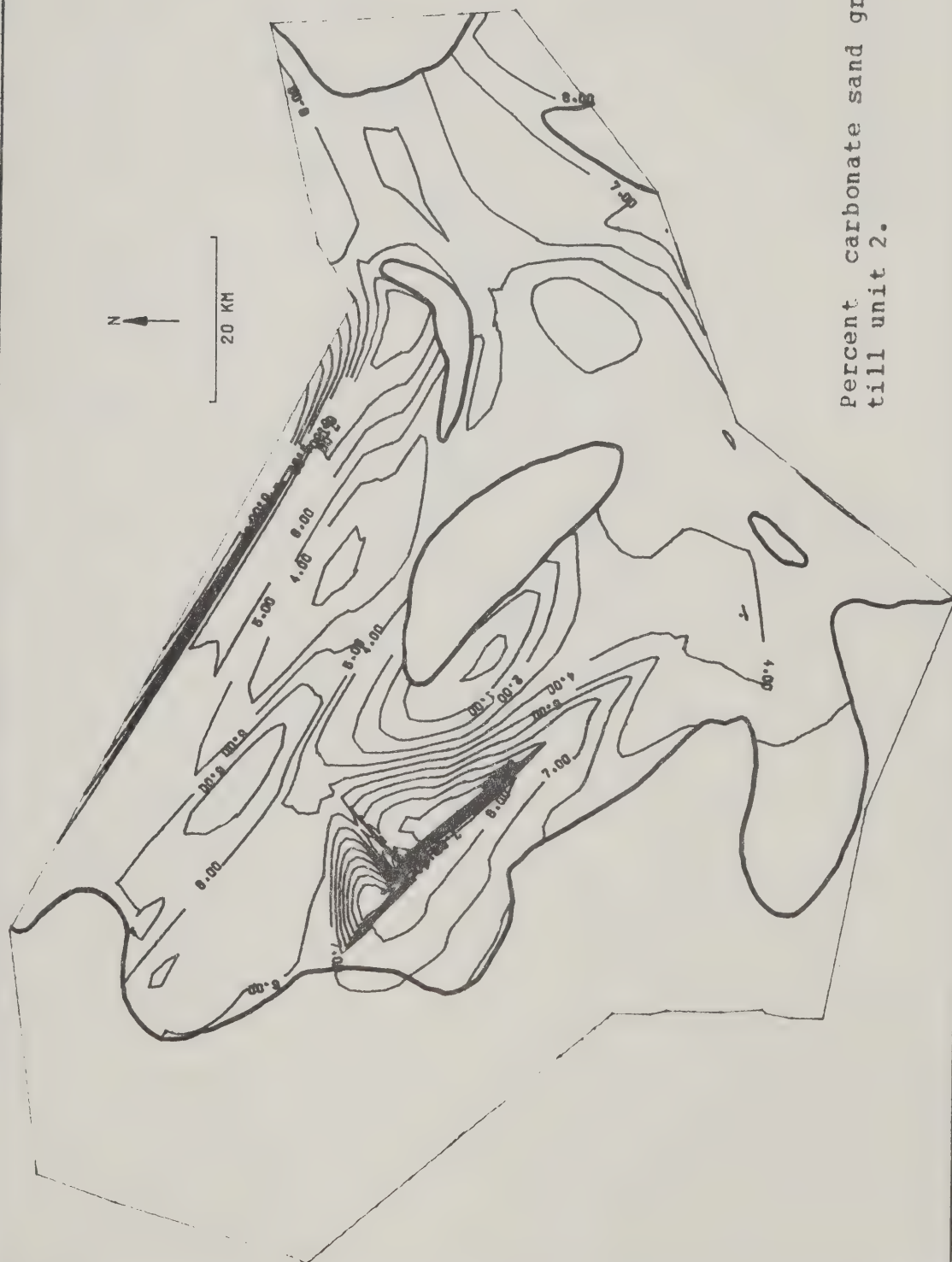


percent basic igneous sand grains  
in till unit 2.



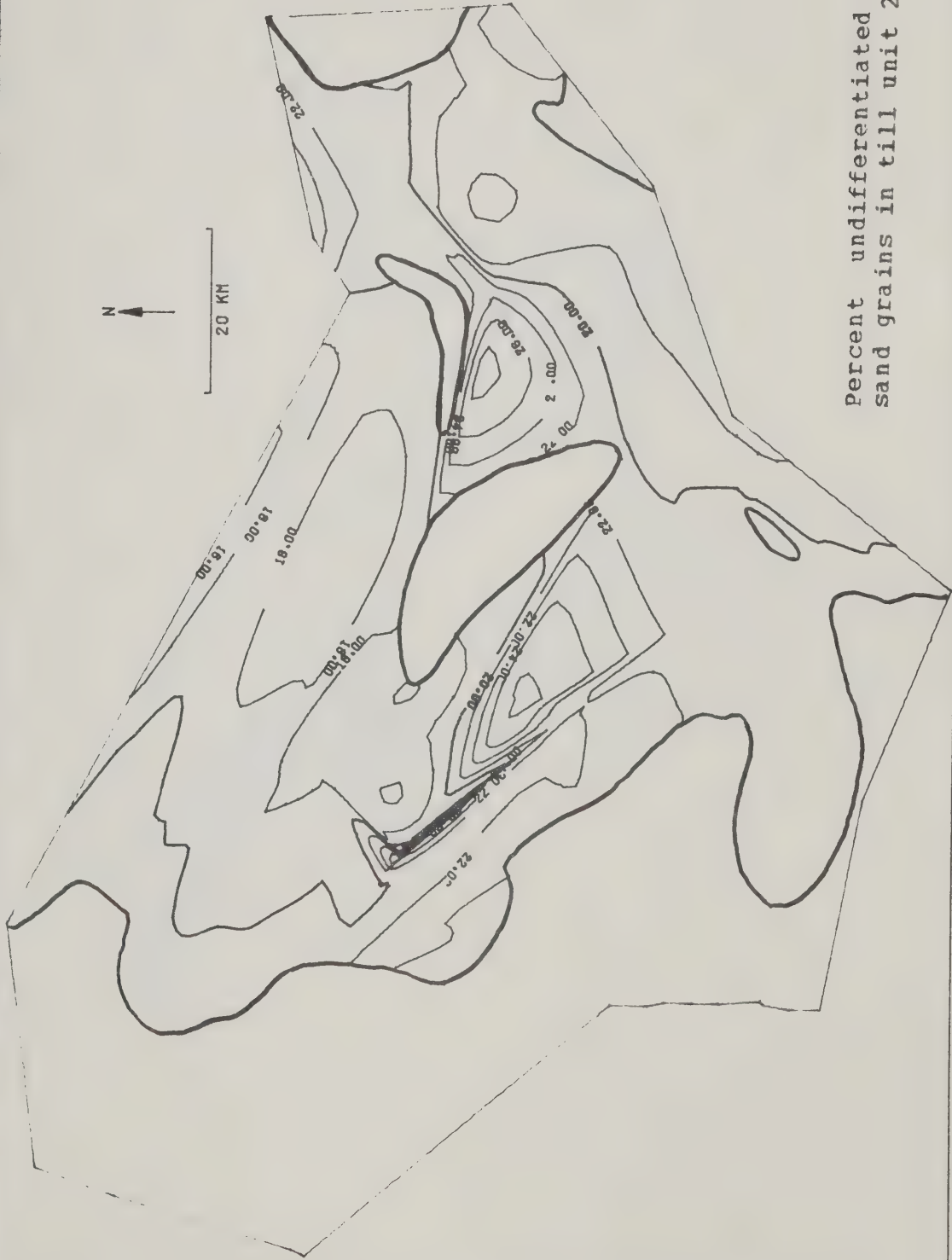


Percent carbonate sand grains in  
till unit 2.

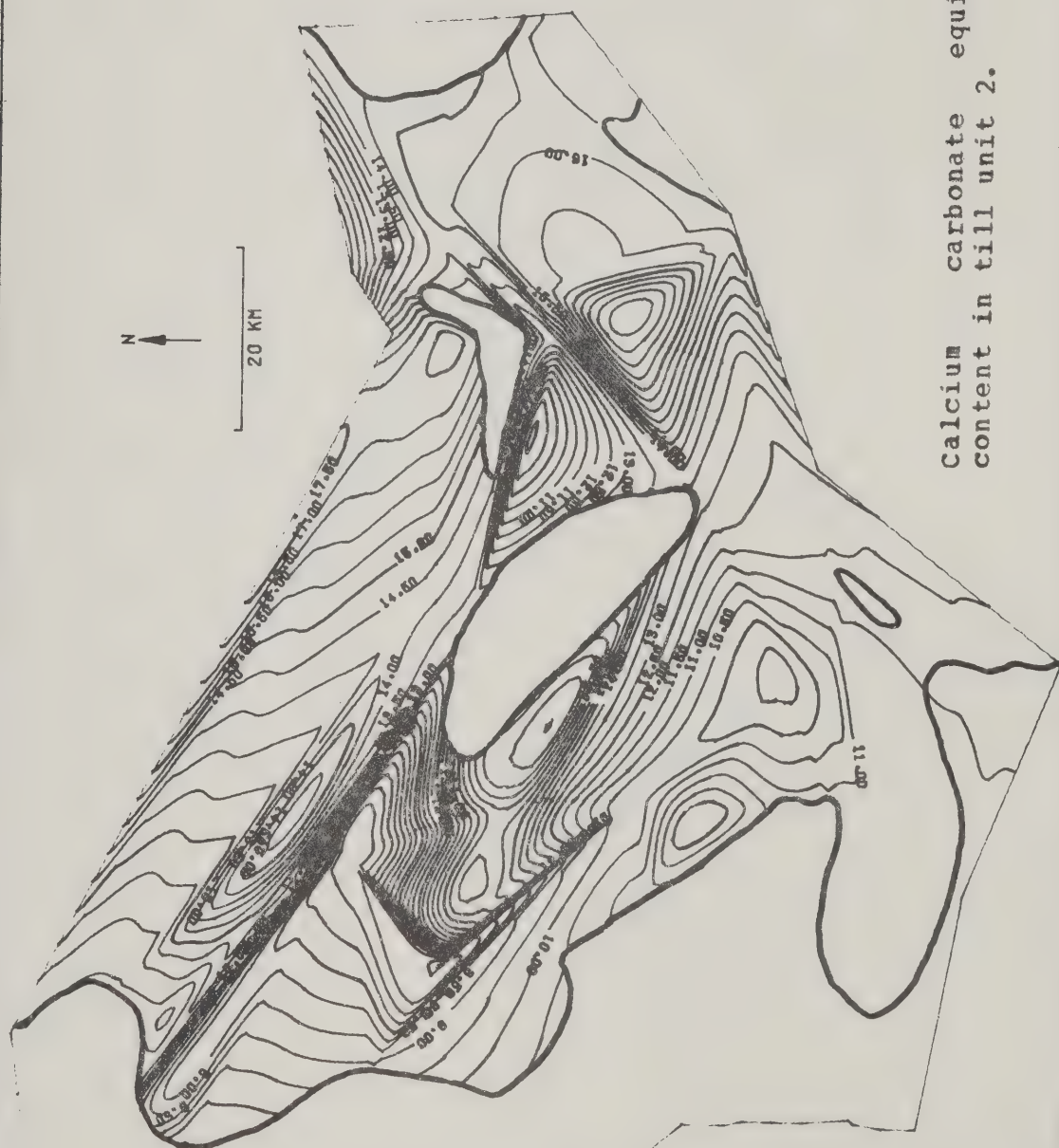




Percent undifferentiated quartz  
sand grains in till unit 2.







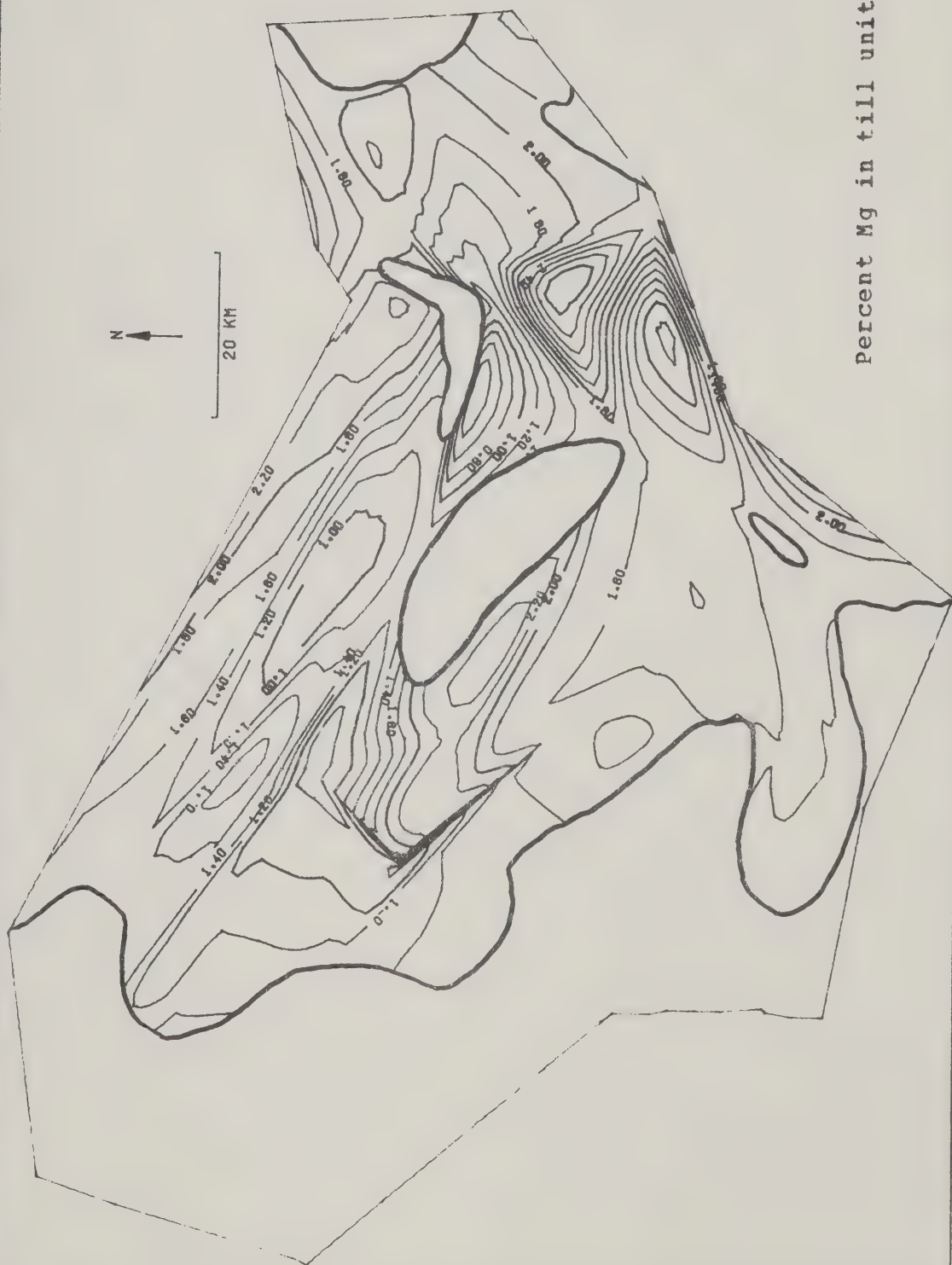
Calcium carbonate equivalent  
content in till unit 2.









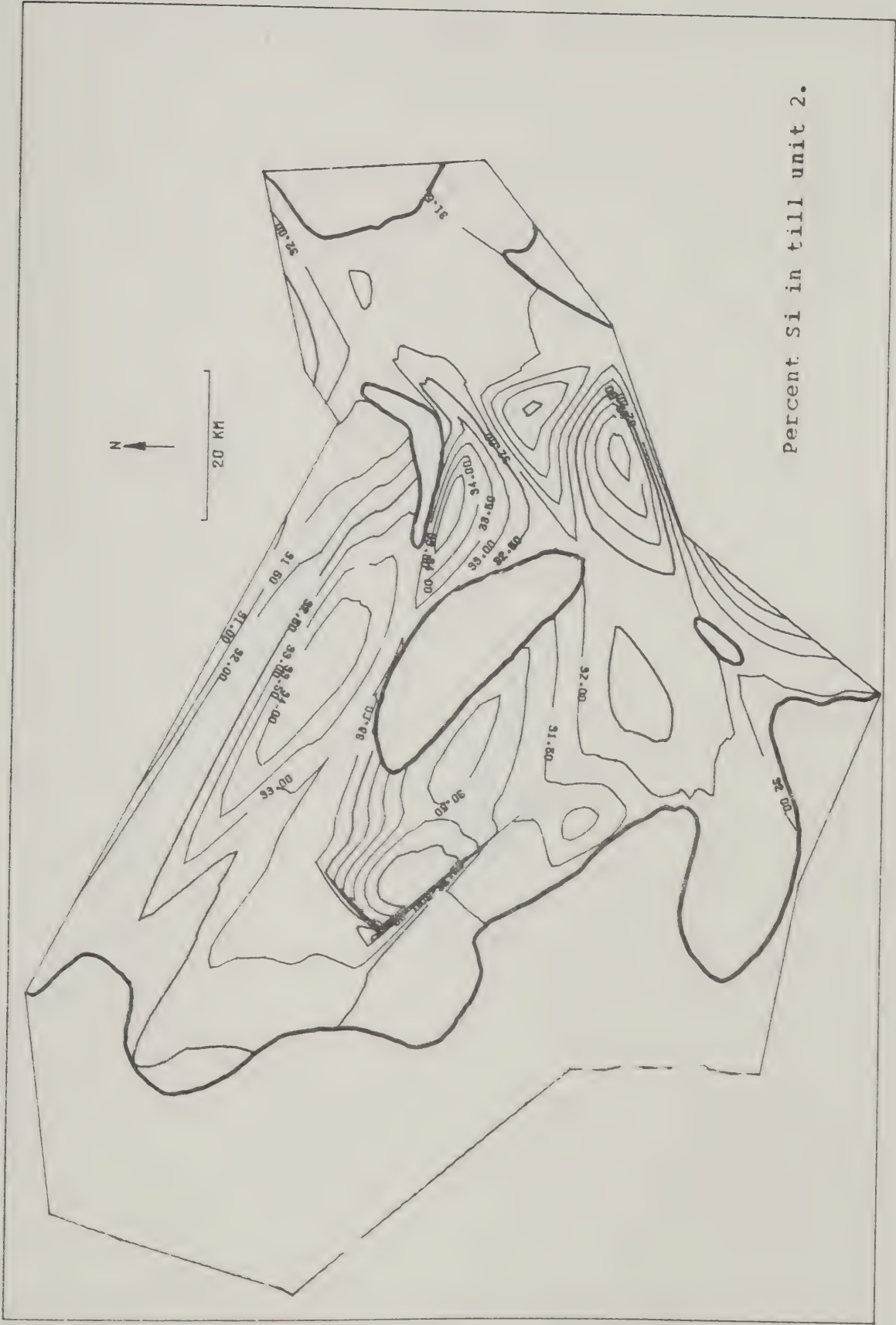


Percent Mg in till unit 2.



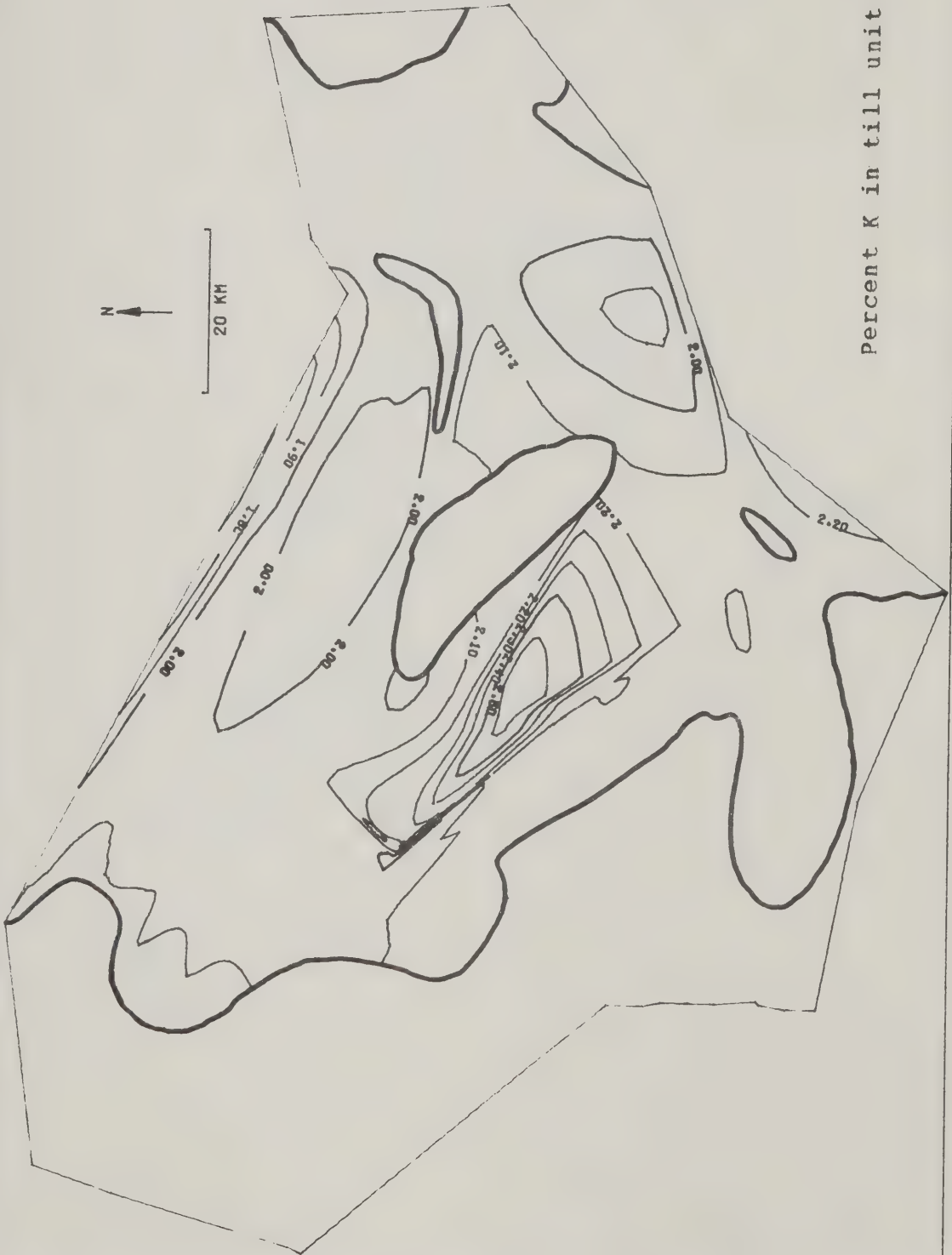






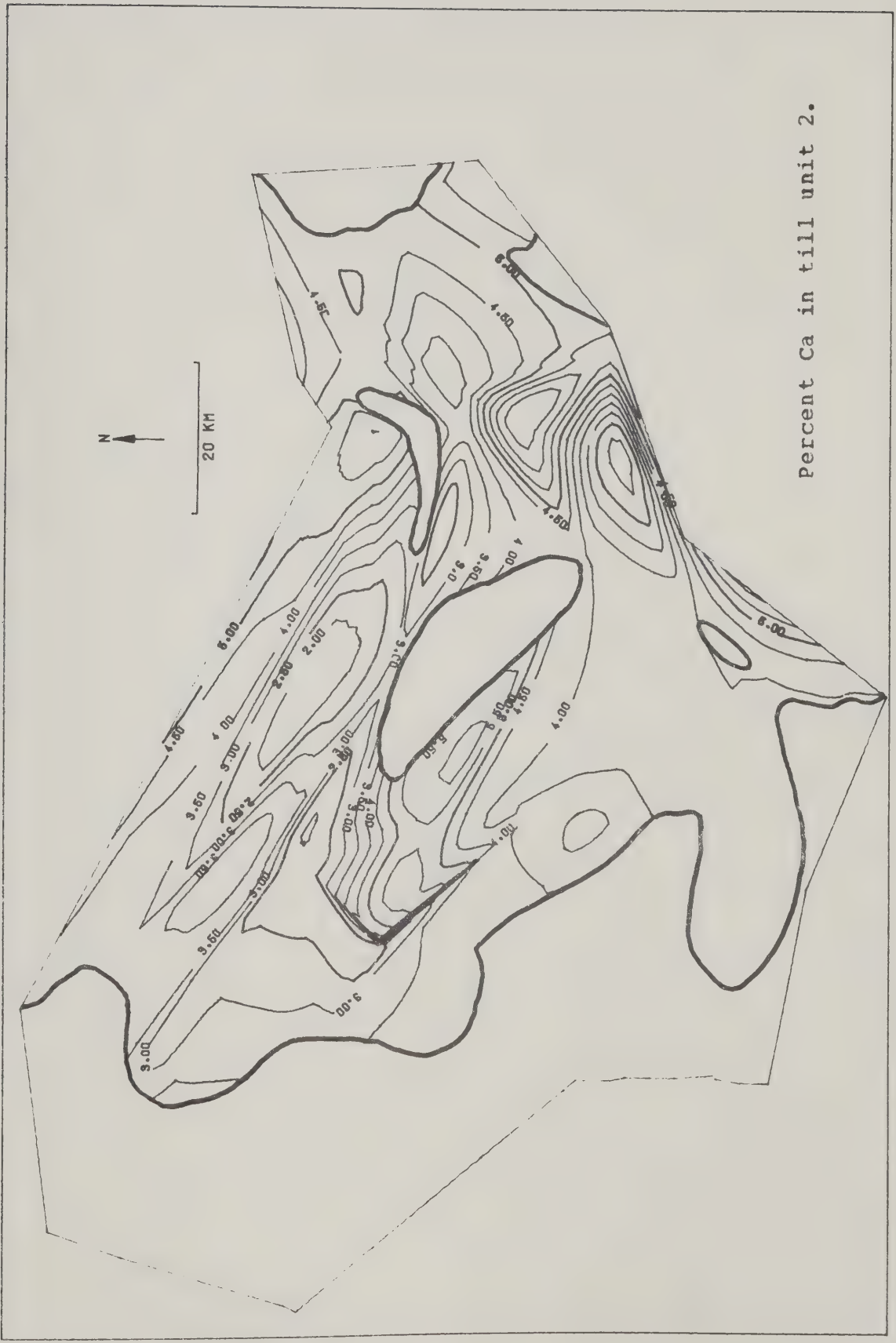






Percent K in till unit 2.



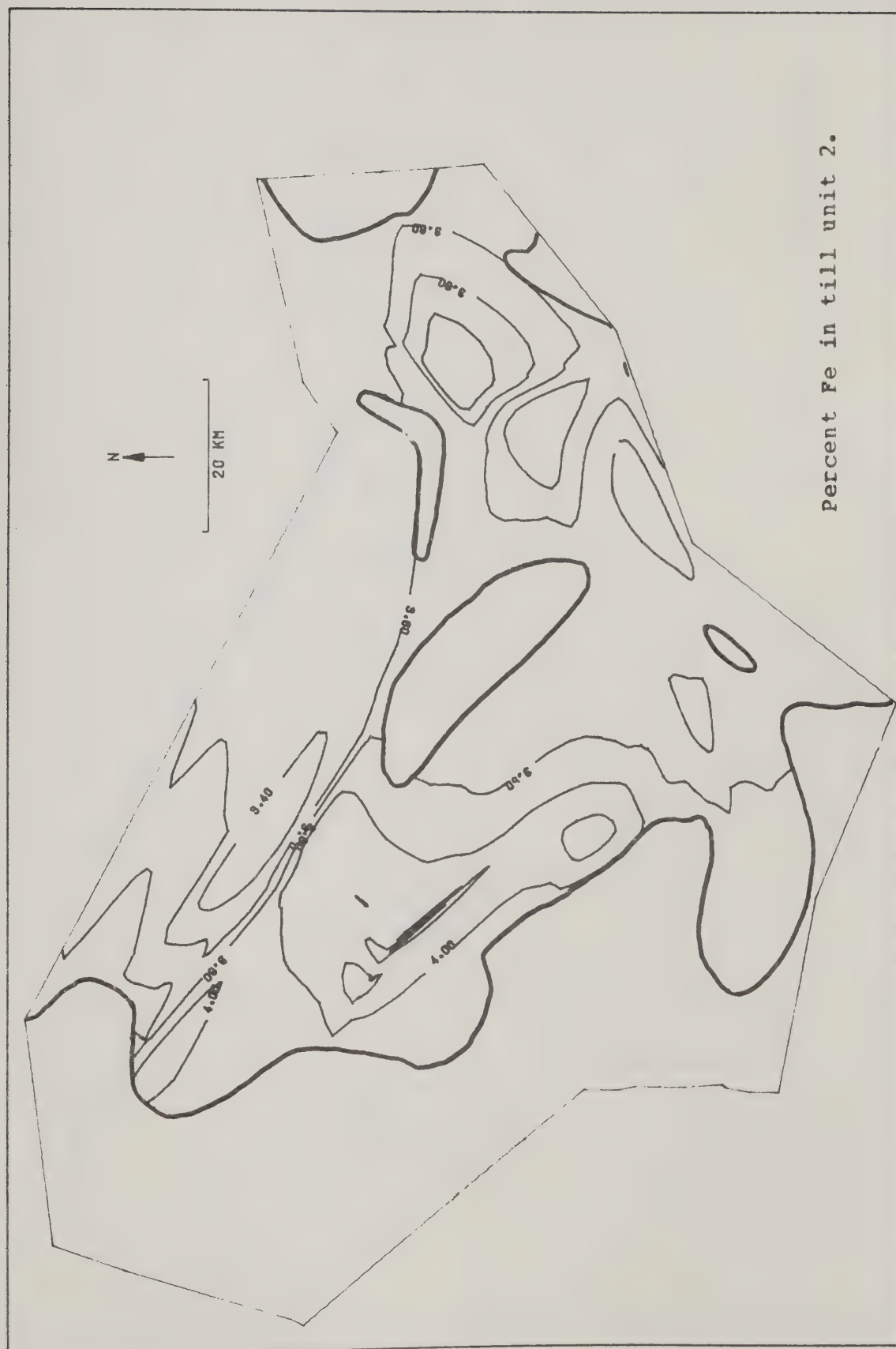


Percent Ca in till unit 2.



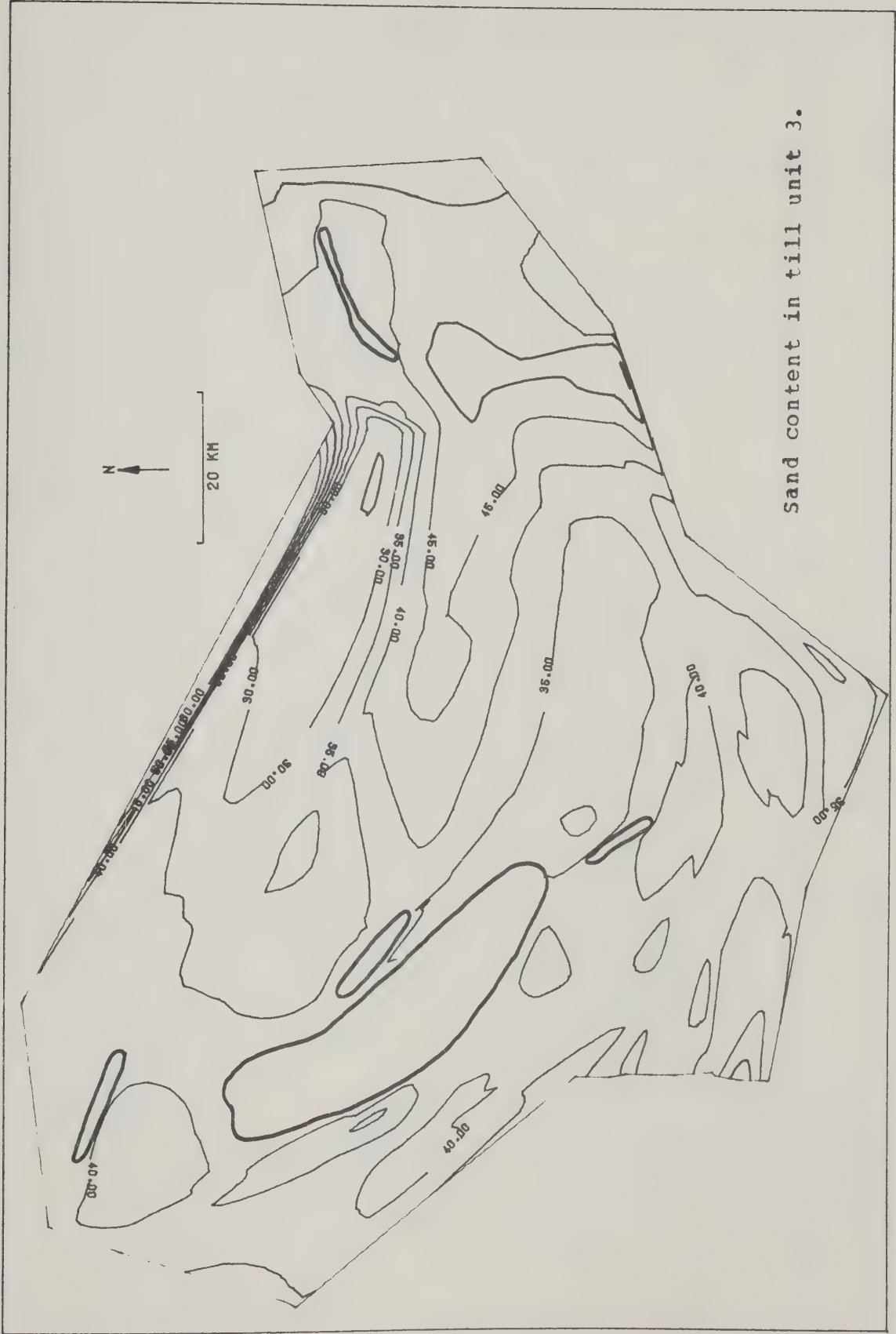








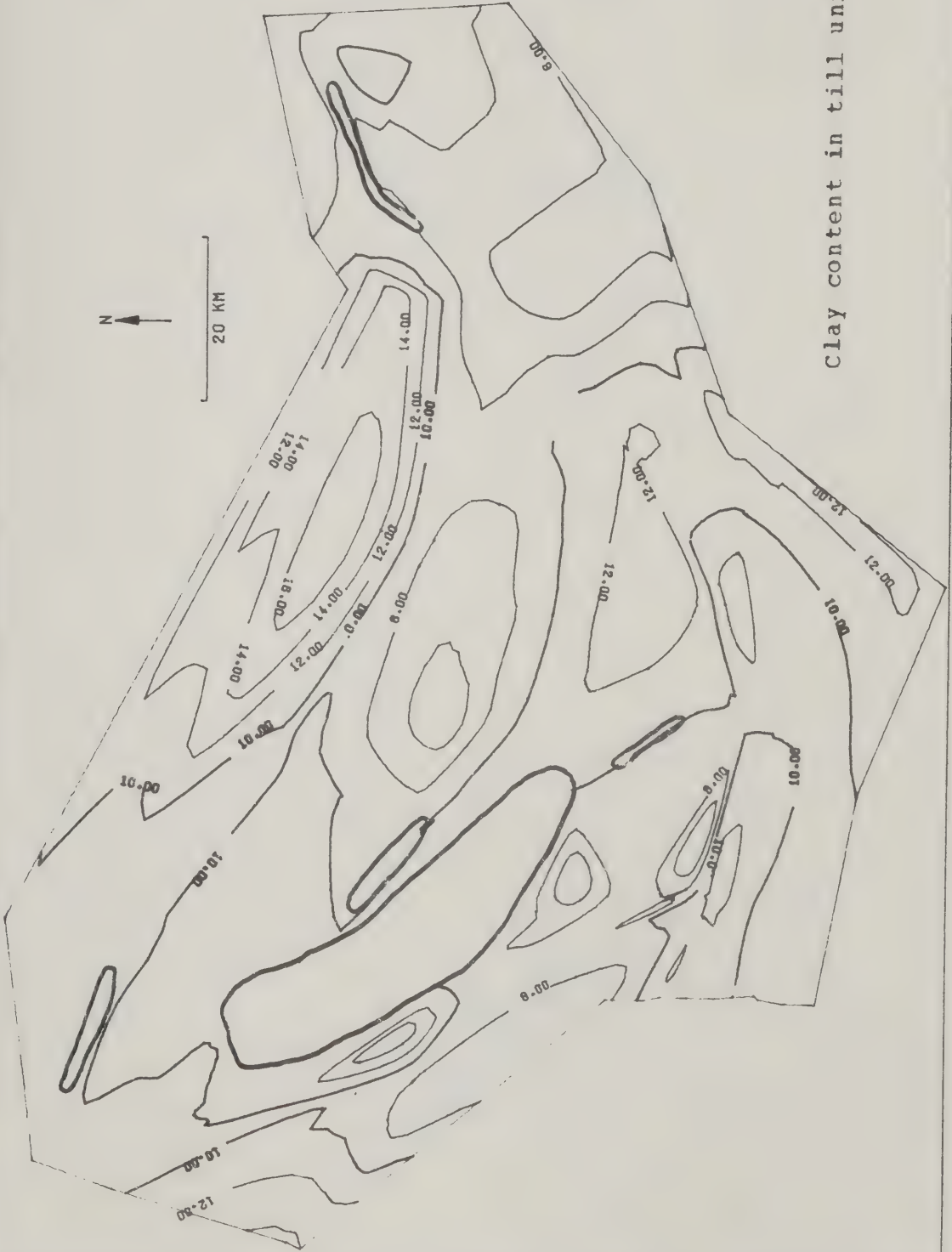




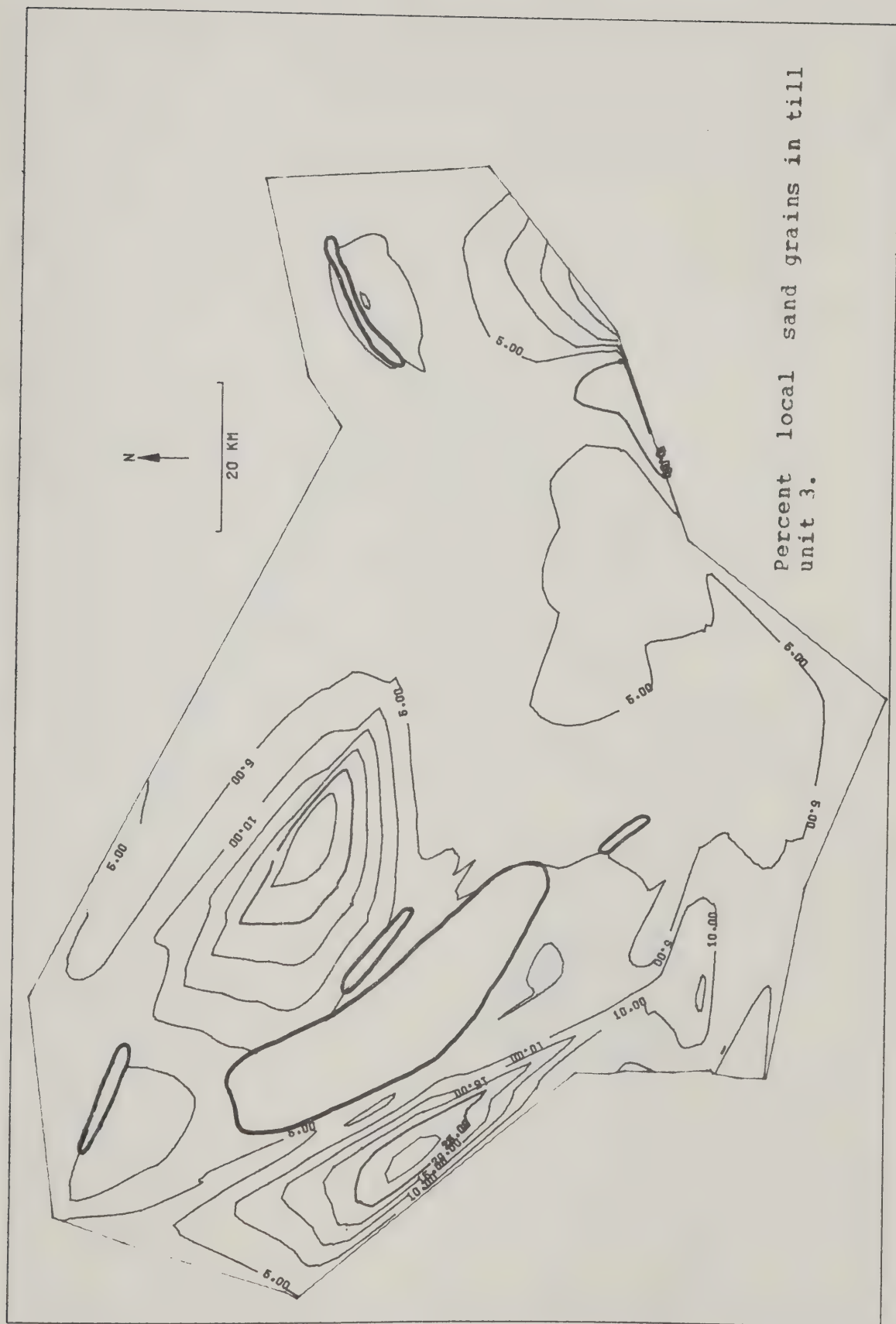
Sand content in till unit 3.



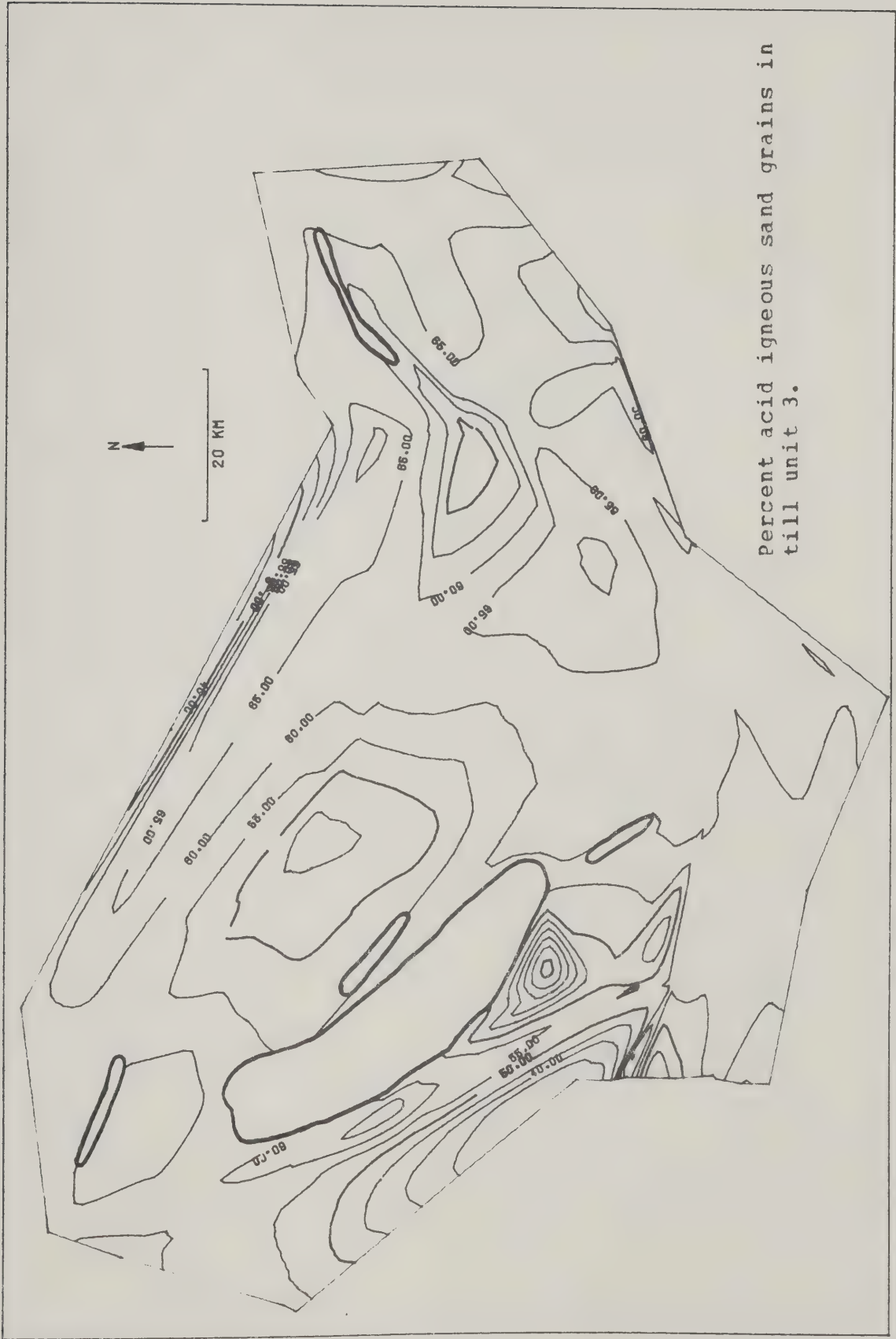
Clay content in till unit 3.







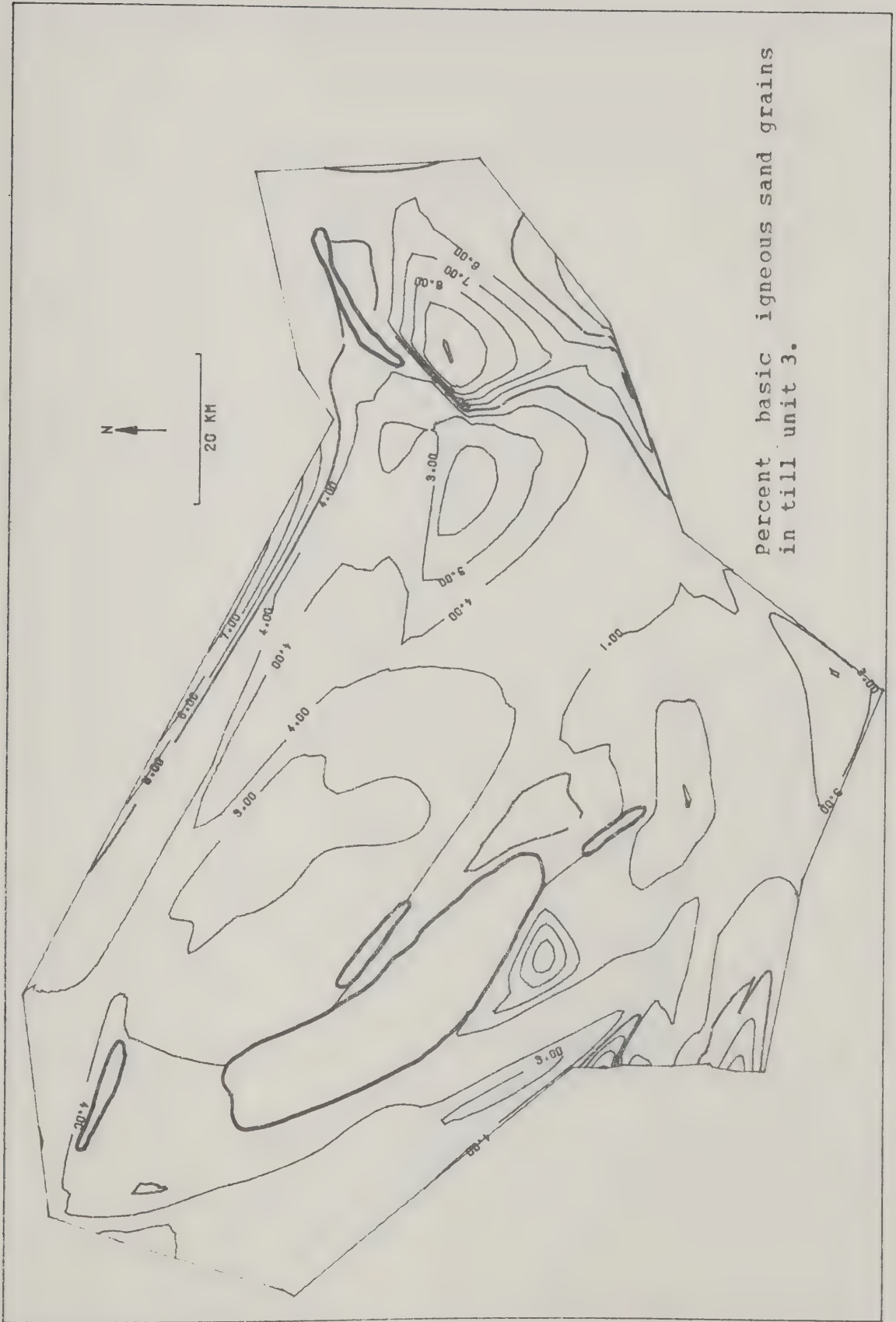




Percent acid igneous sand grains in  
till unit 3.

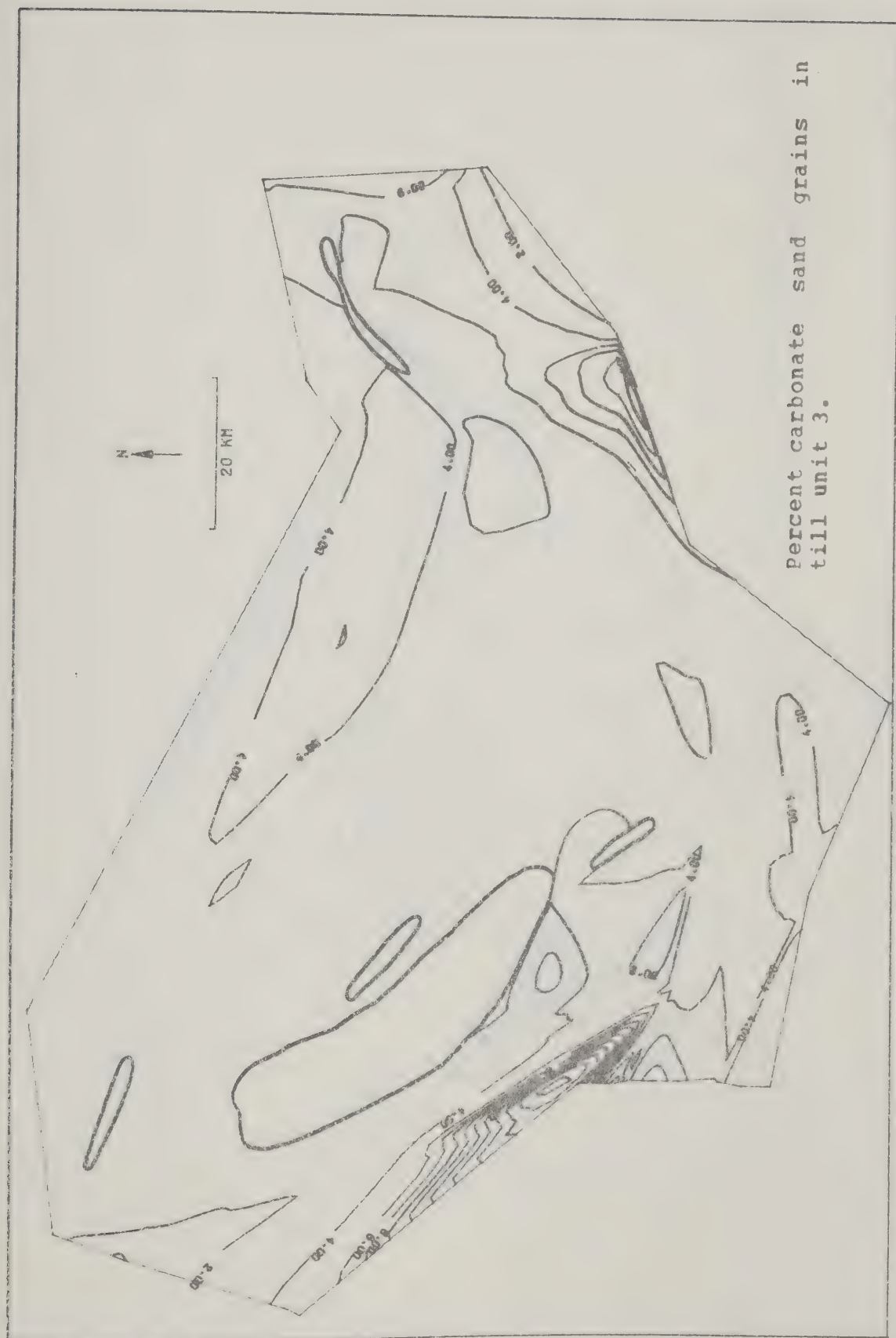




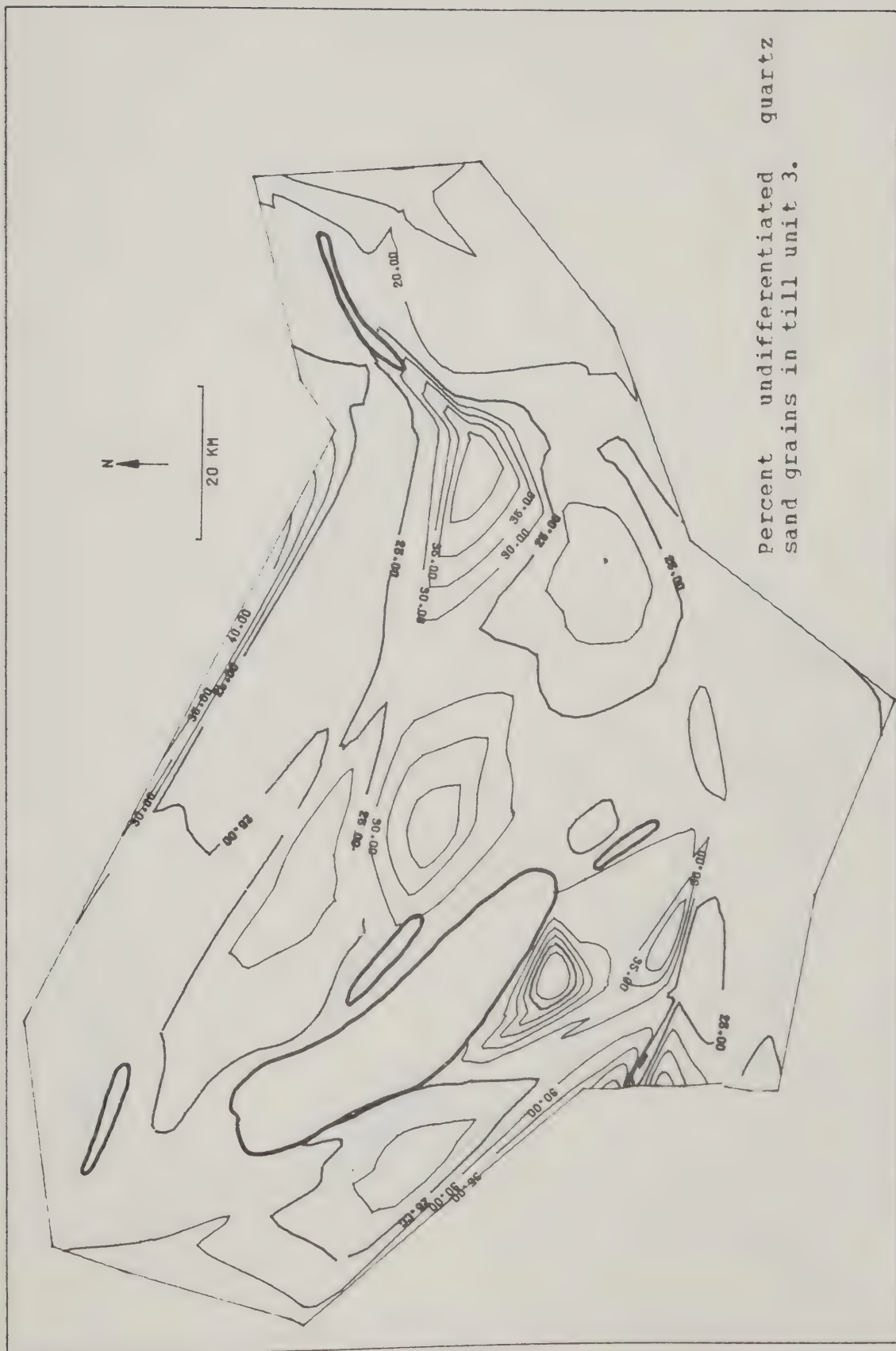


Percent basic igneous sand grains  
in till unit 3.









Percent undifferentiated quartz sand grains in till unit 3.

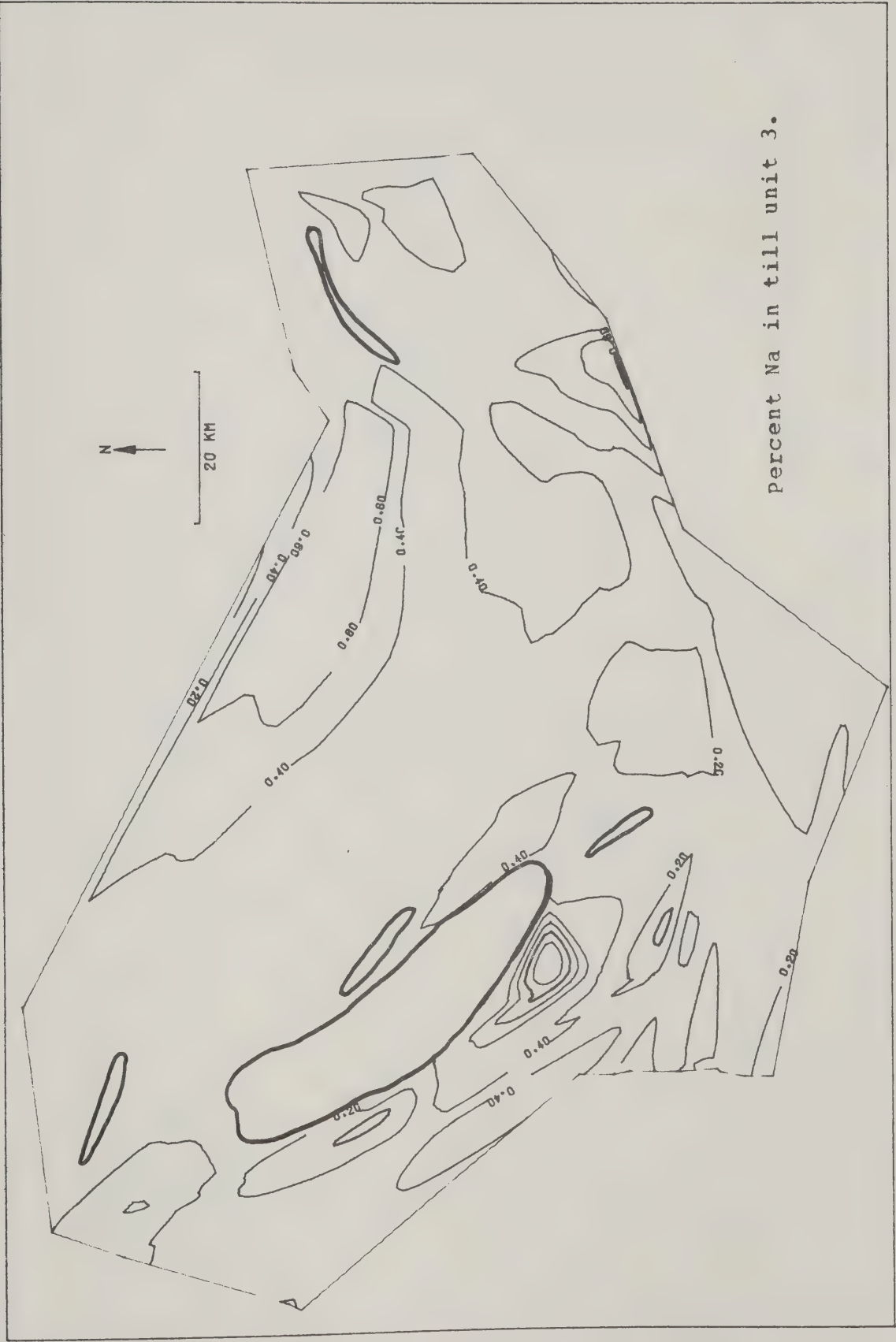




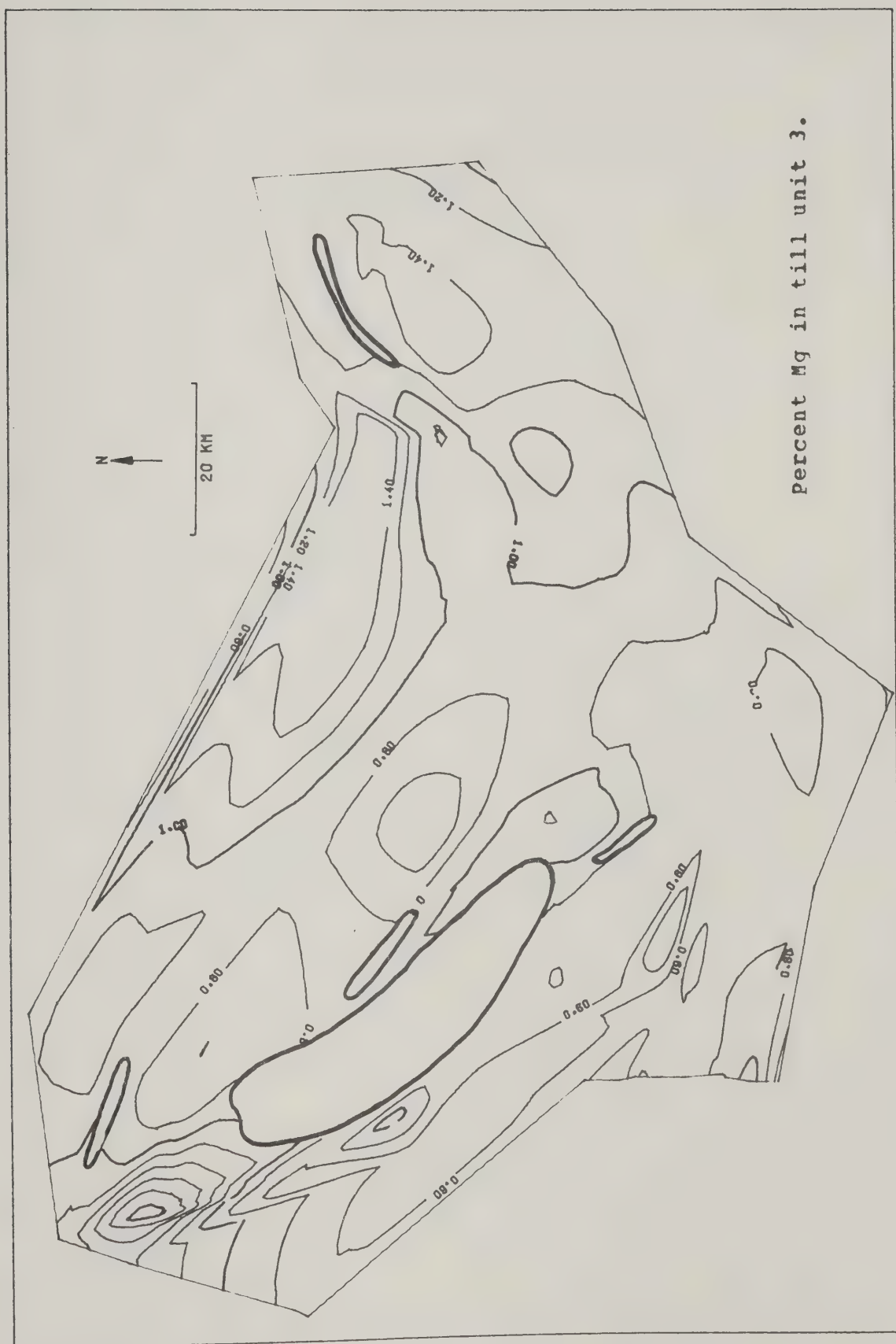
Calcium carbonate equivalent content in till unit 3.





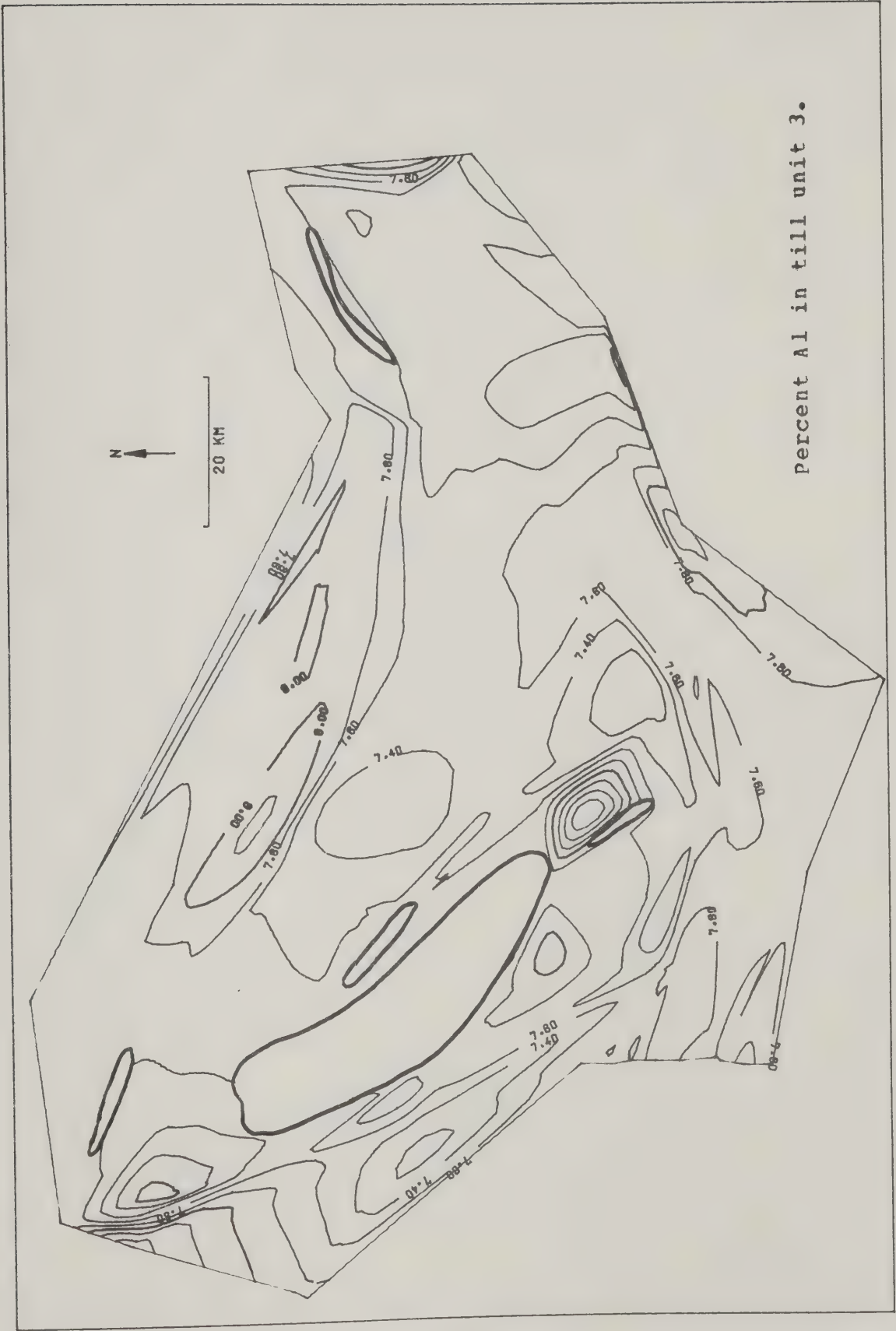






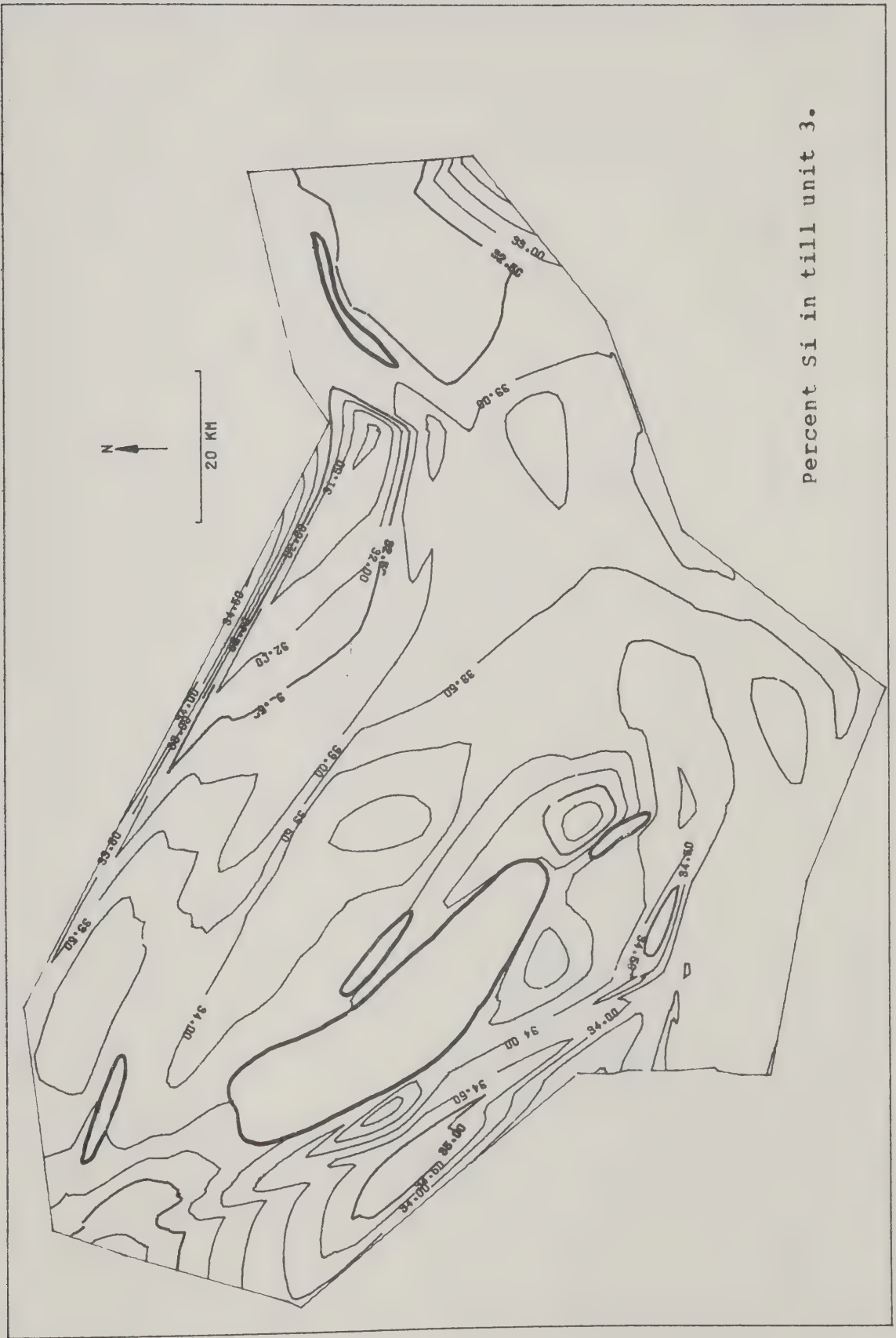
Percent Mg in till unit 3.





Percent Al in till unit 3.





Percent Si in till unit 3.



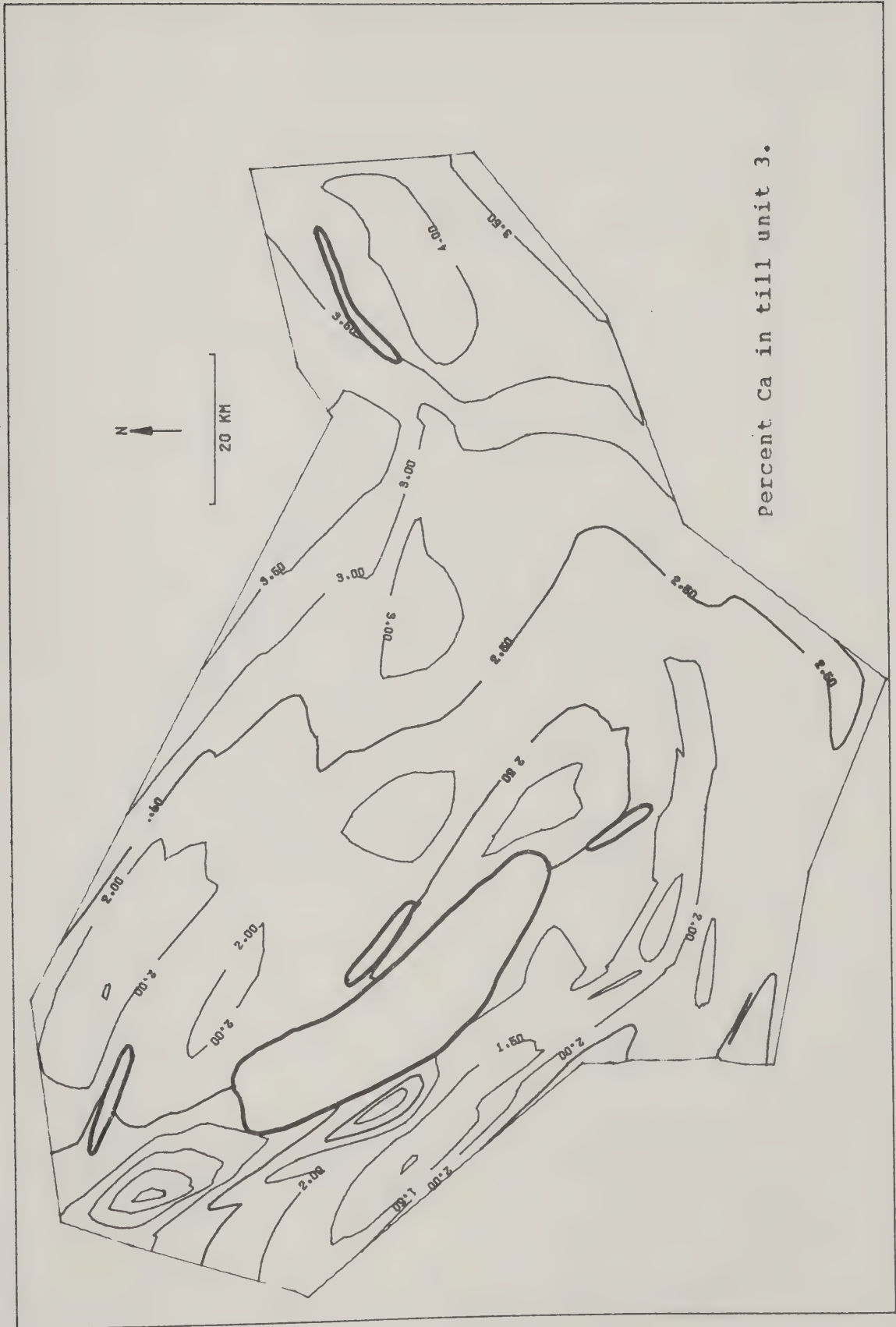




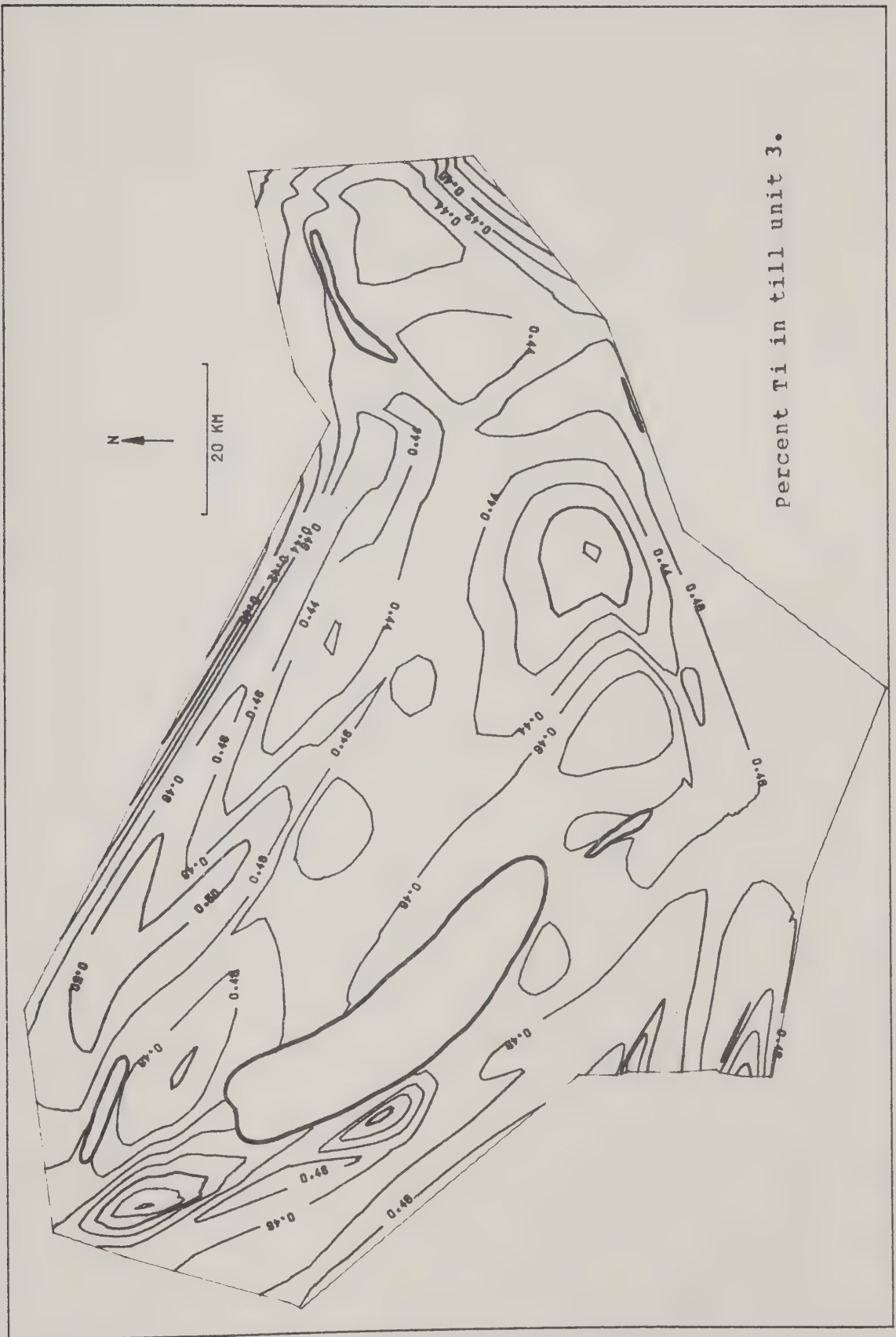
Percent K in till unit 3.



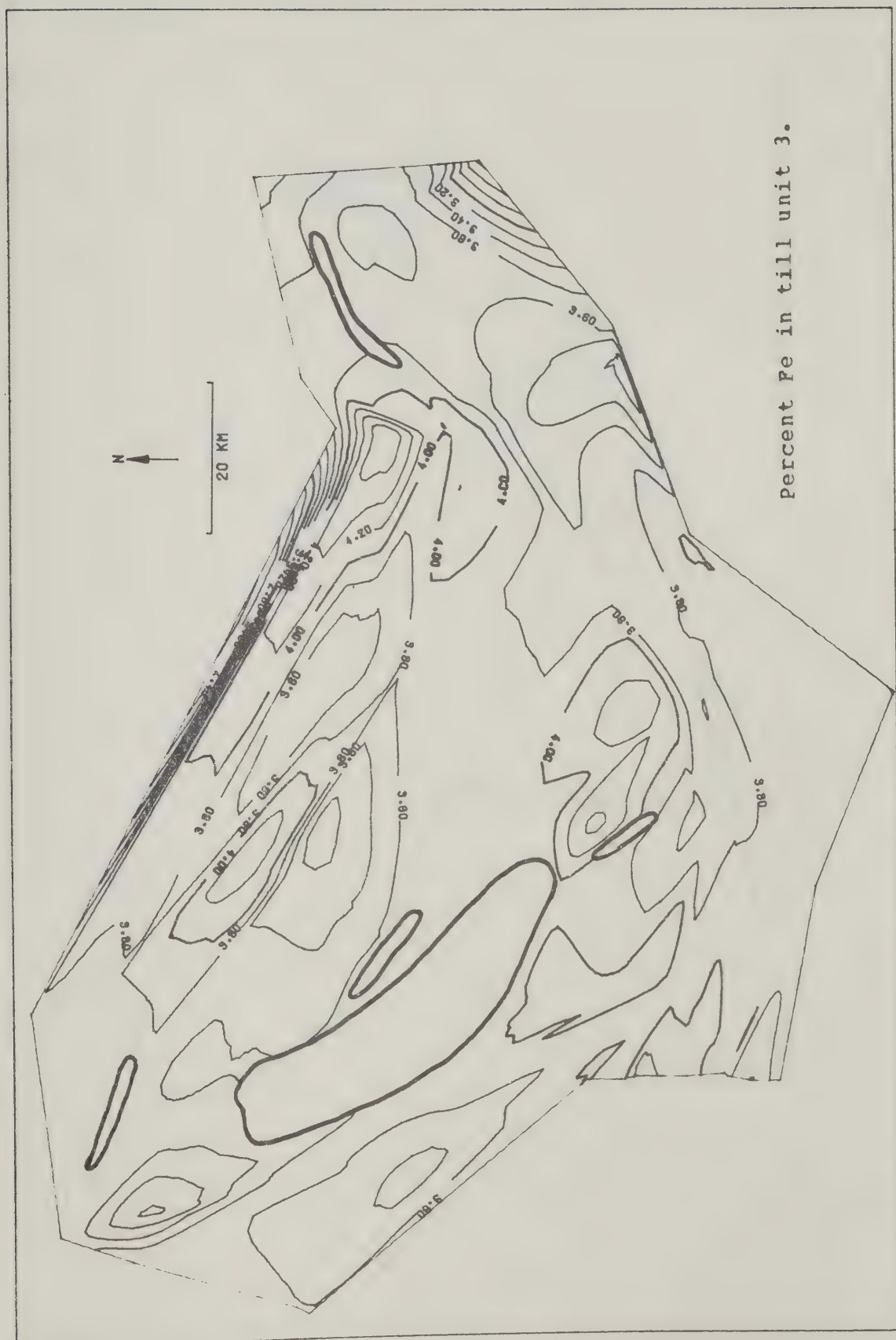
Percent Ca in till unit 3.





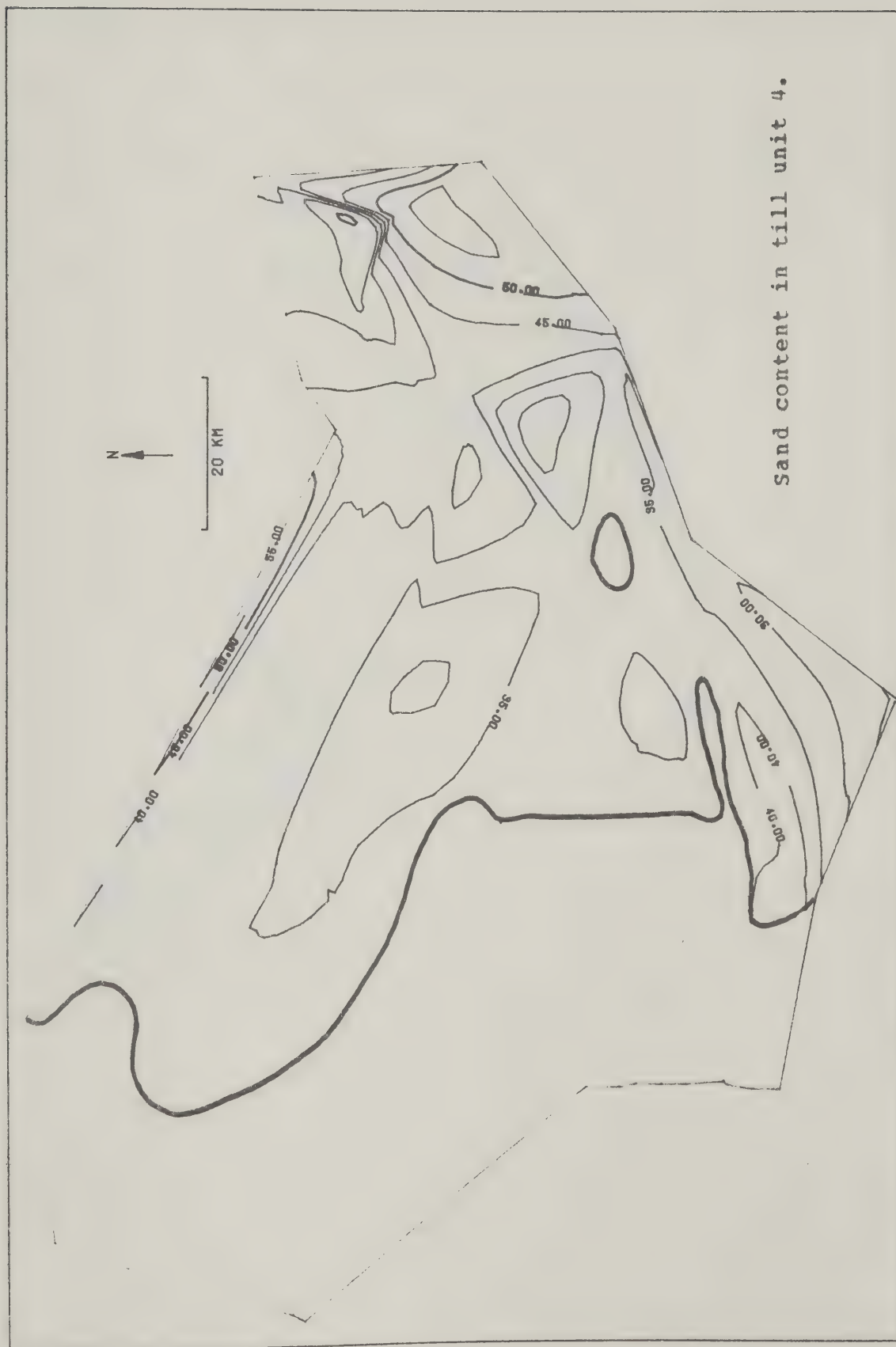




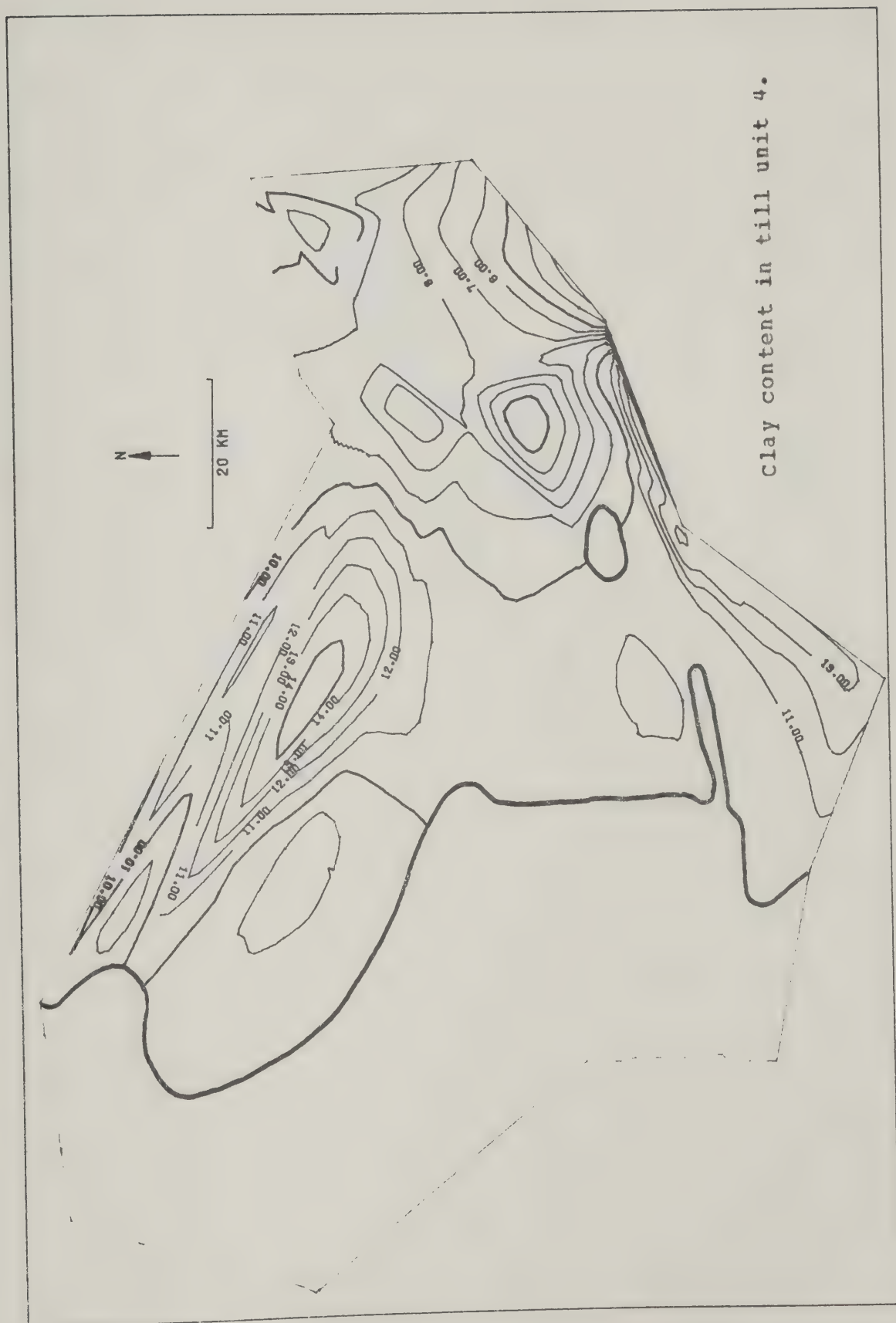




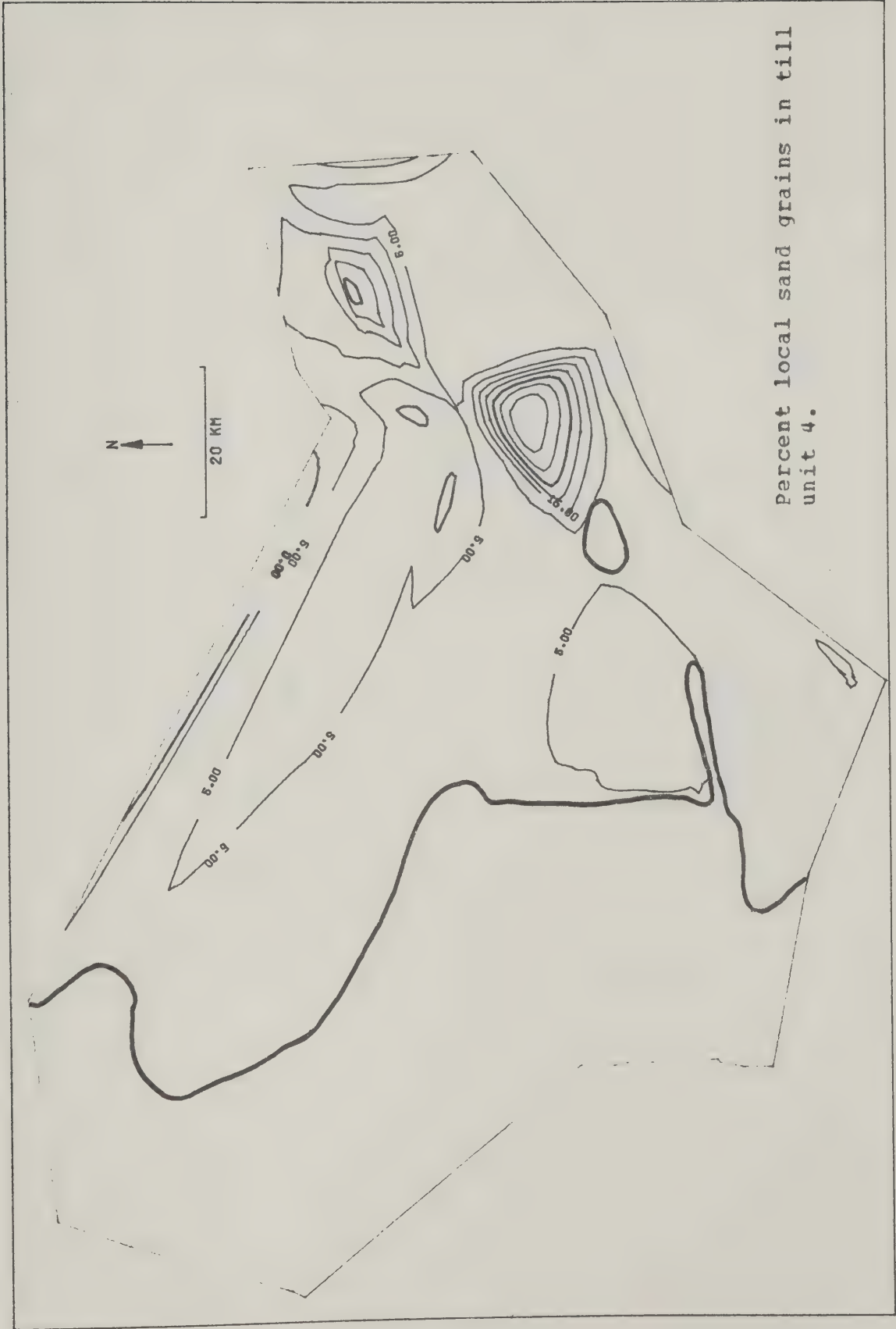






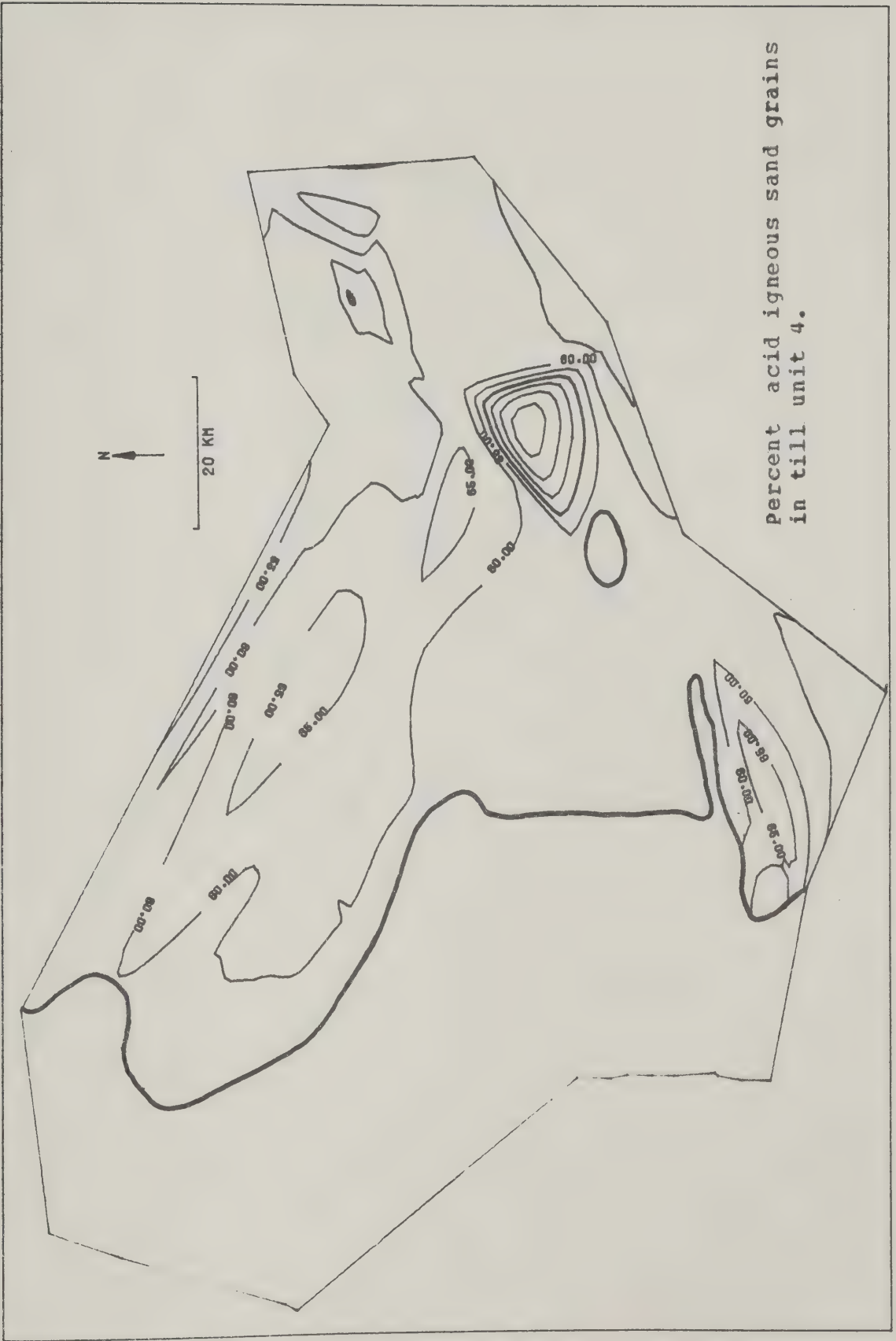






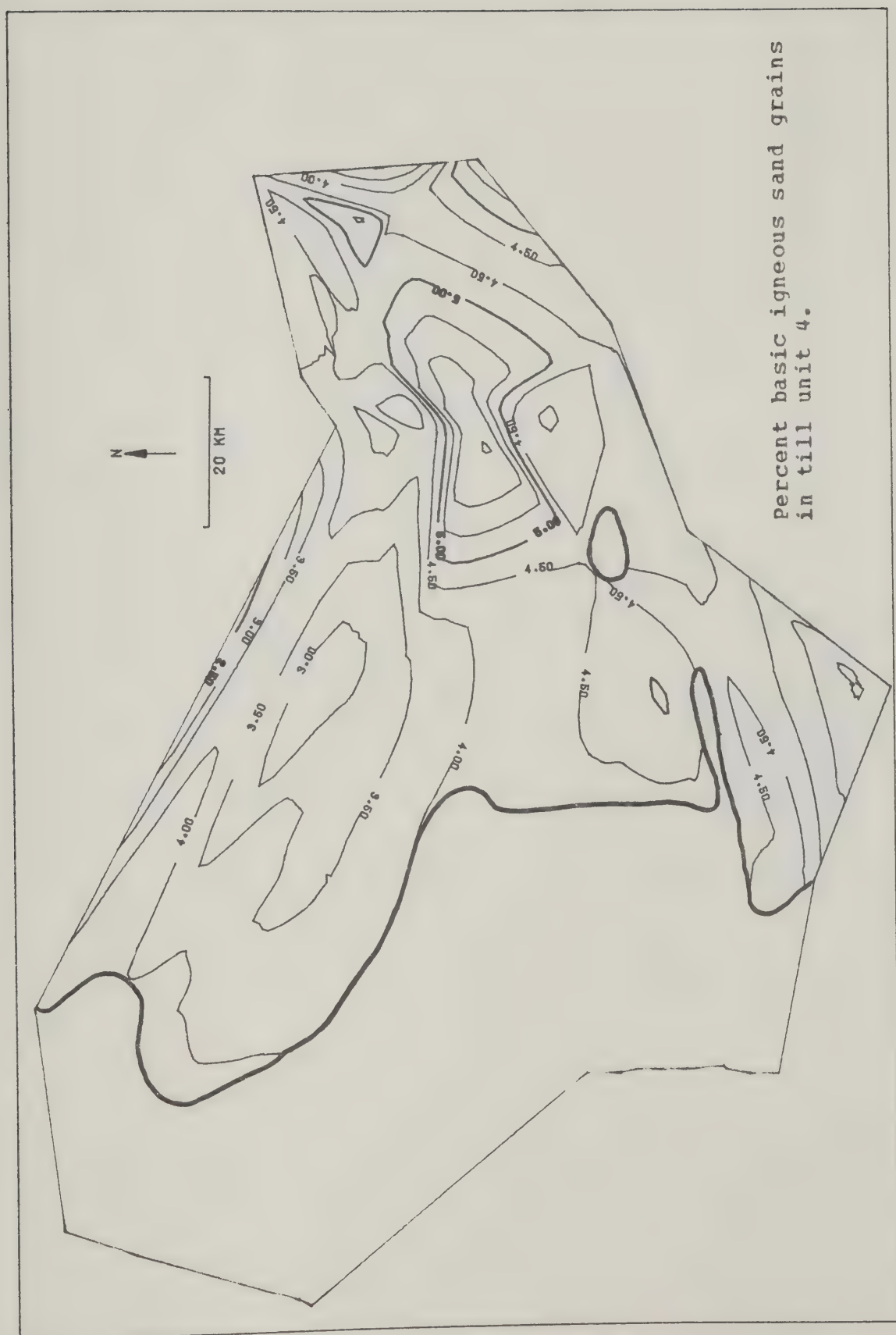
Percent local sand grains in till  
unit 4.





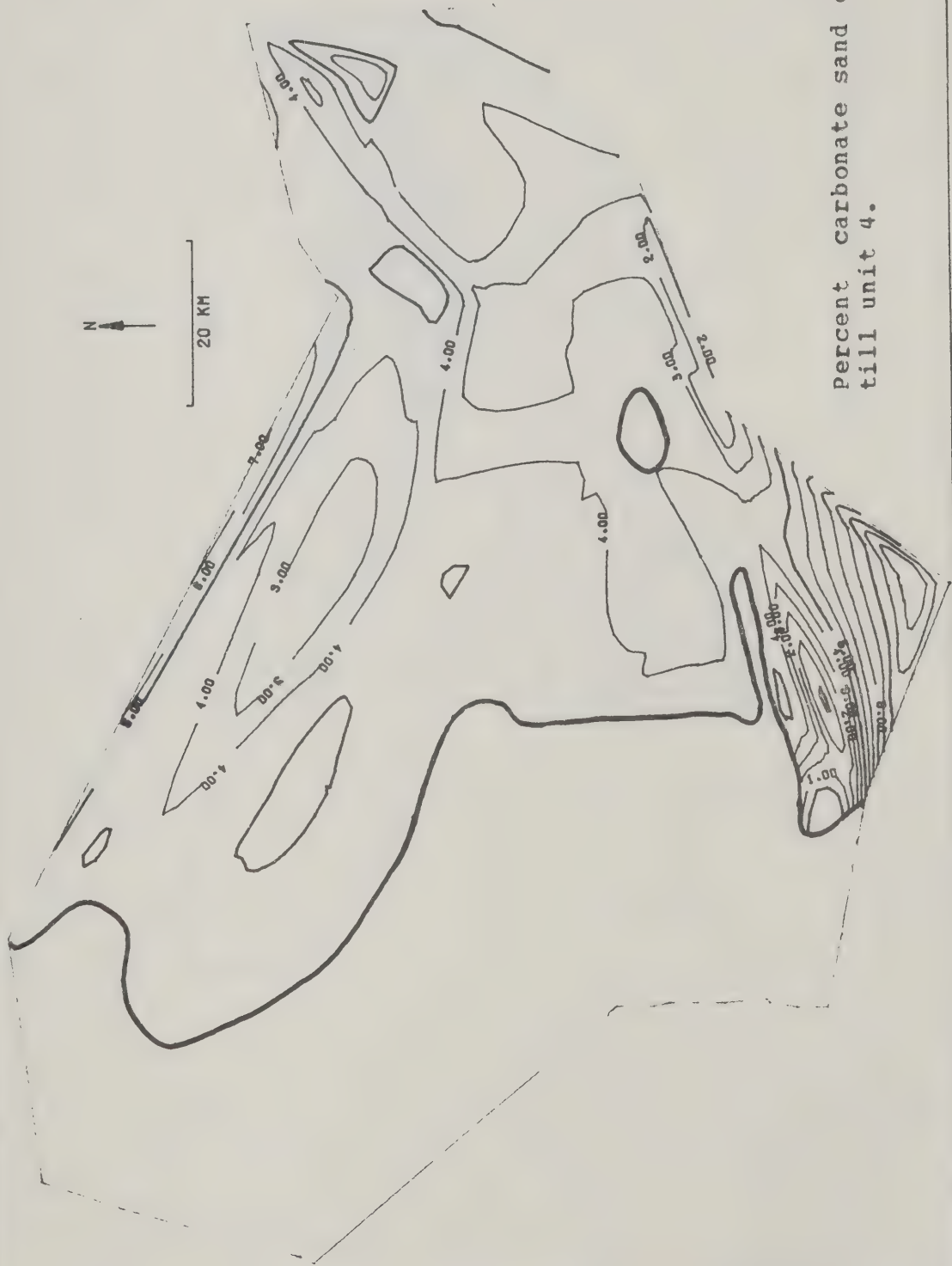






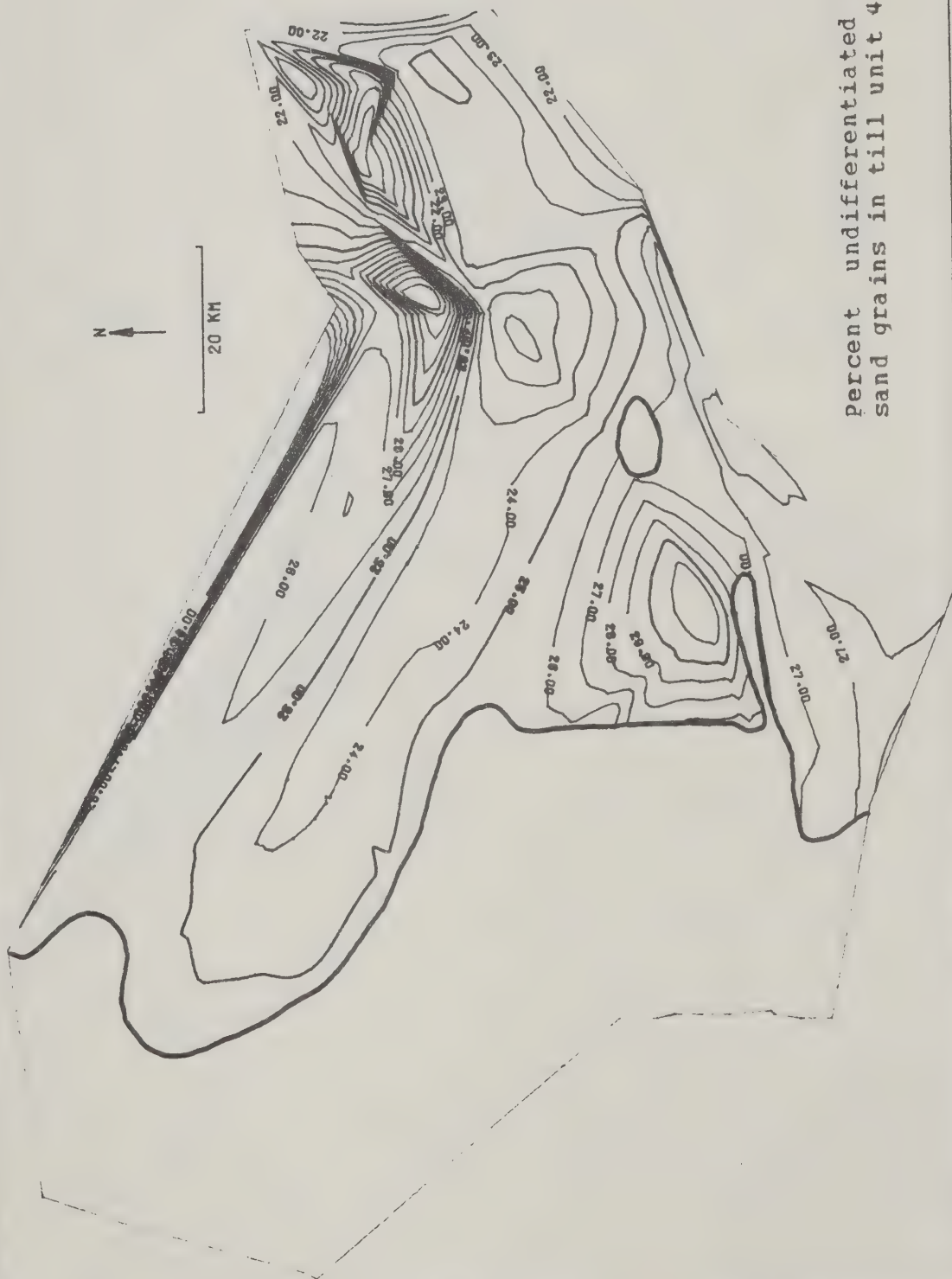


Percent carbonate sand grains in  
till unit 4.

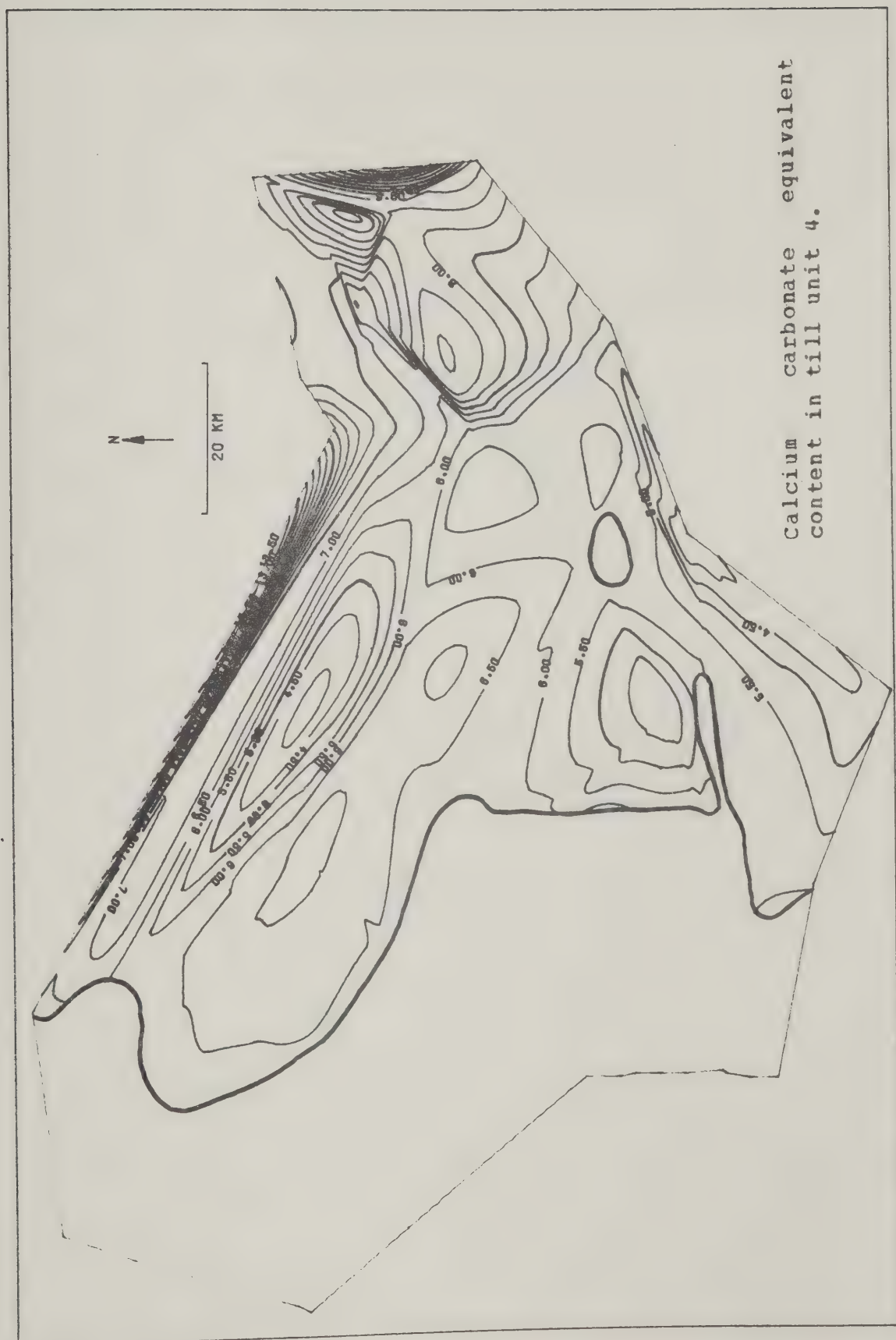




Percent undifferentiated quartz  
sand grains in till unit 4.



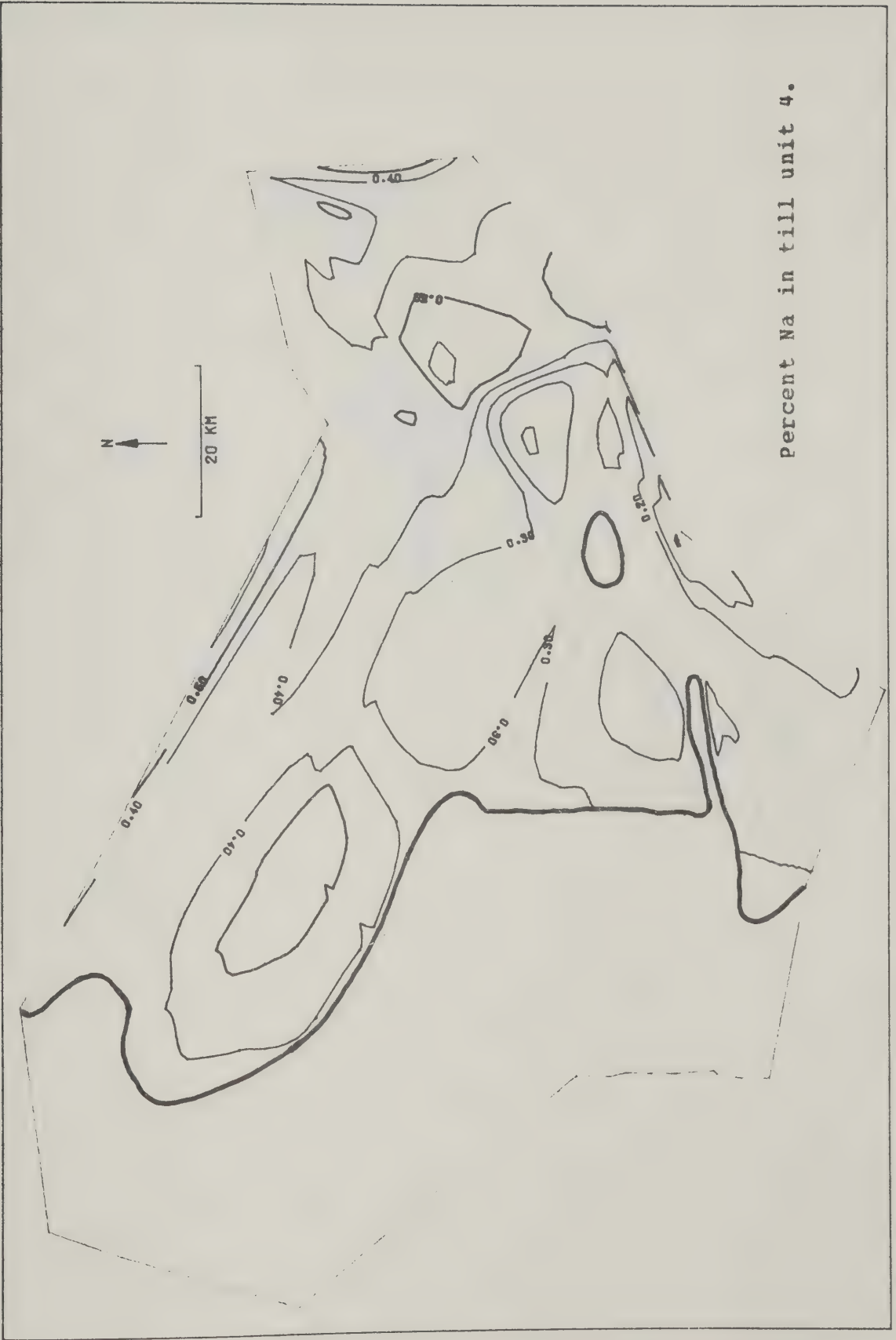




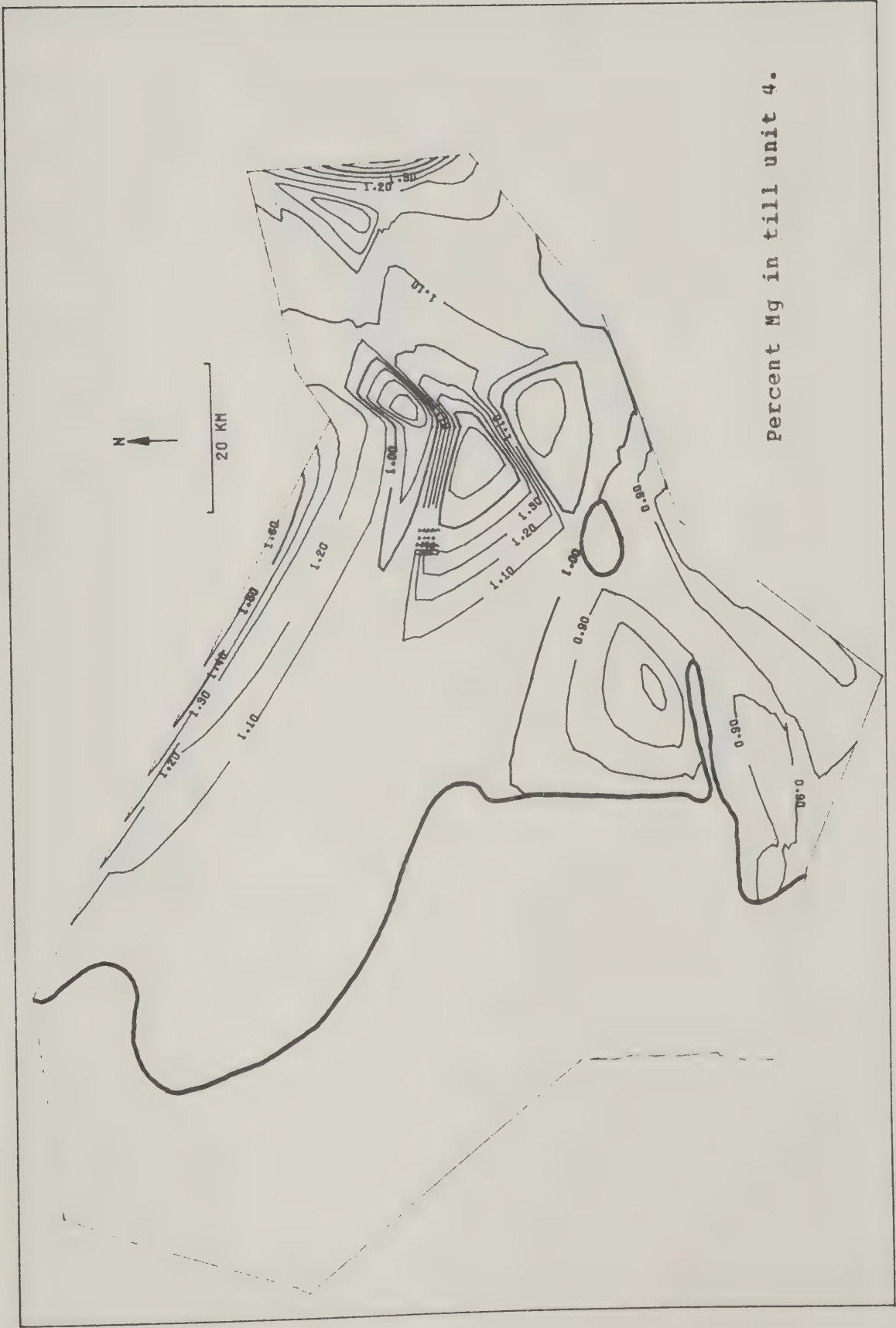




Percent Na in till unit 4.

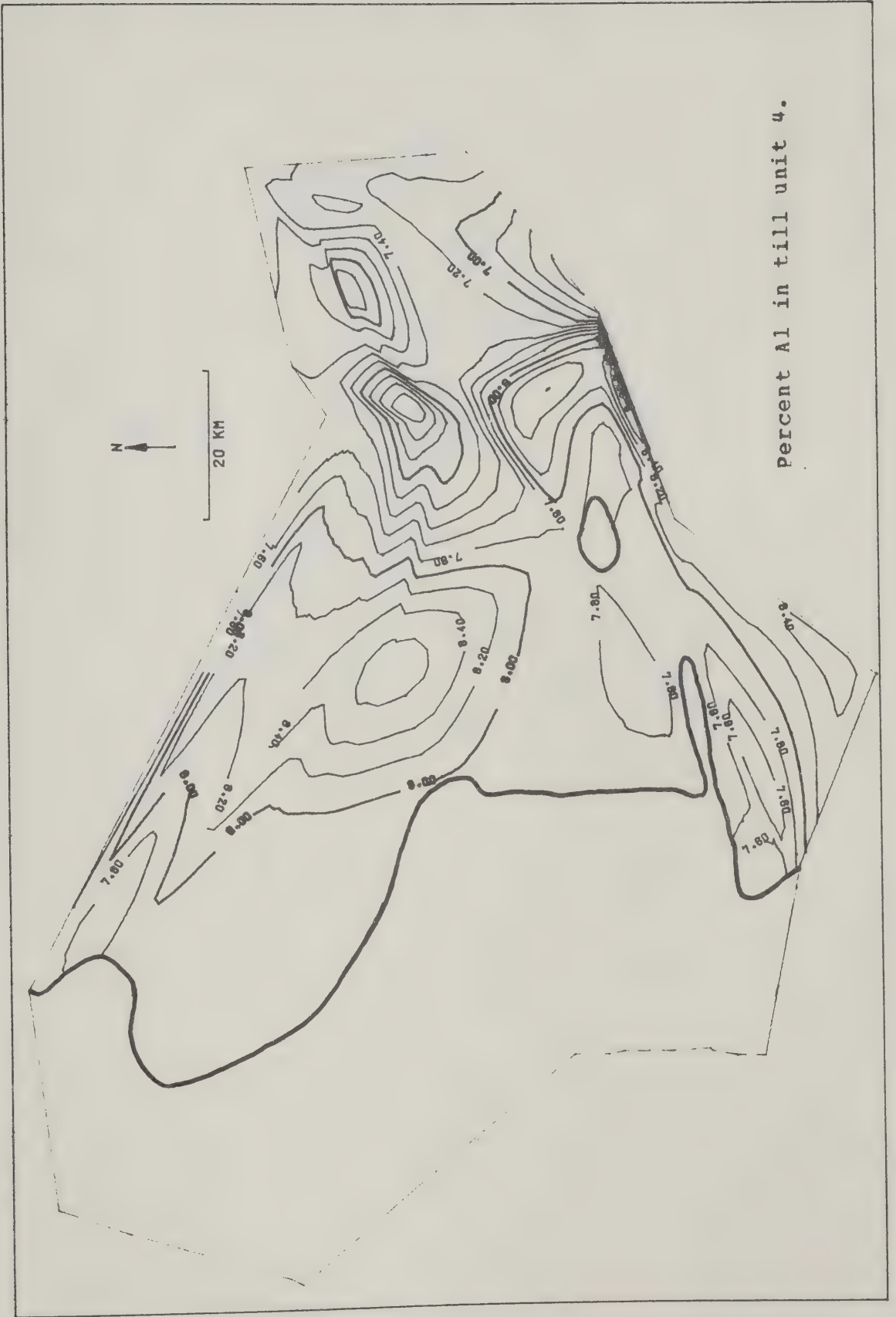




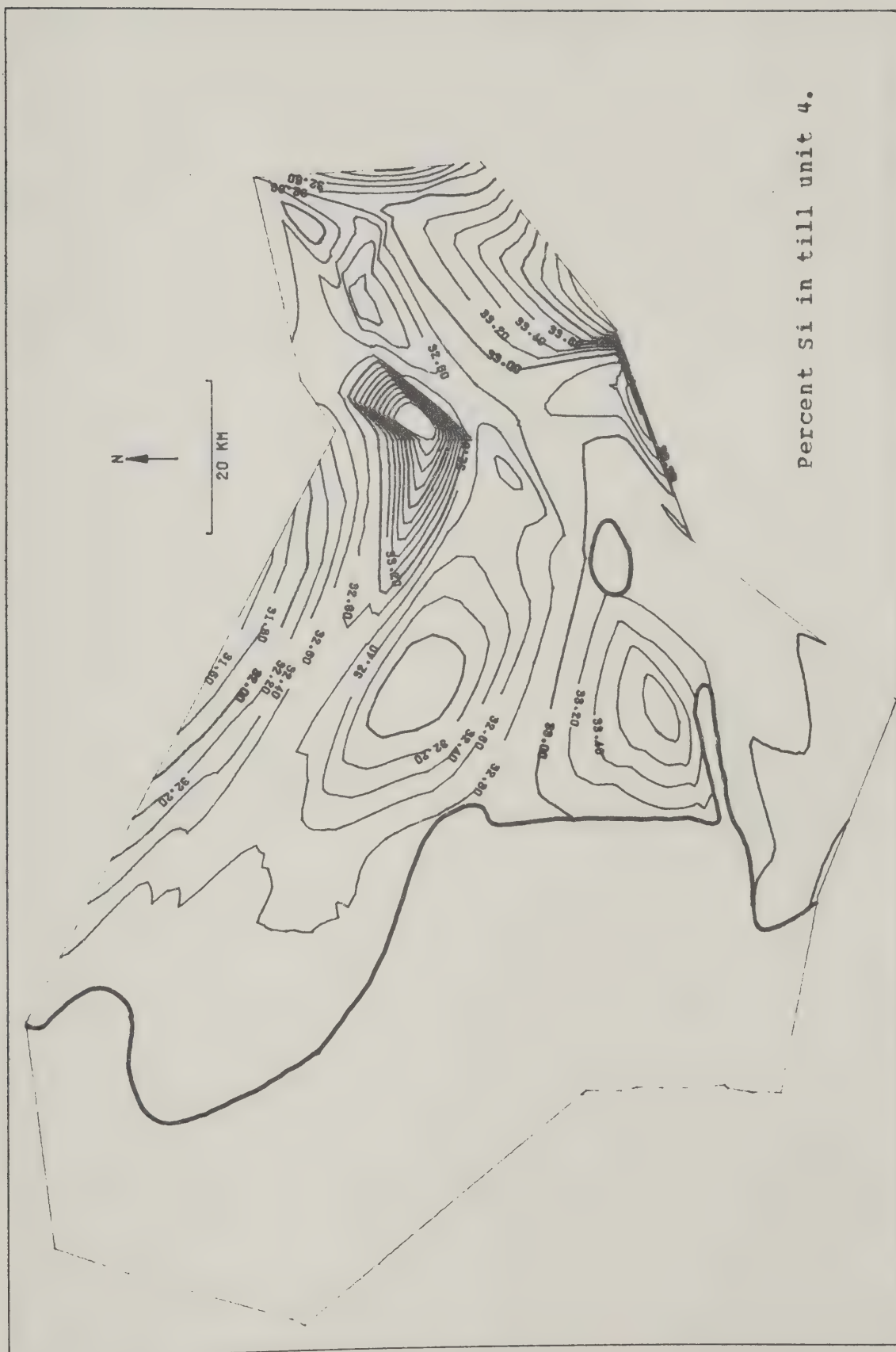




Percent Al in till unit 4.







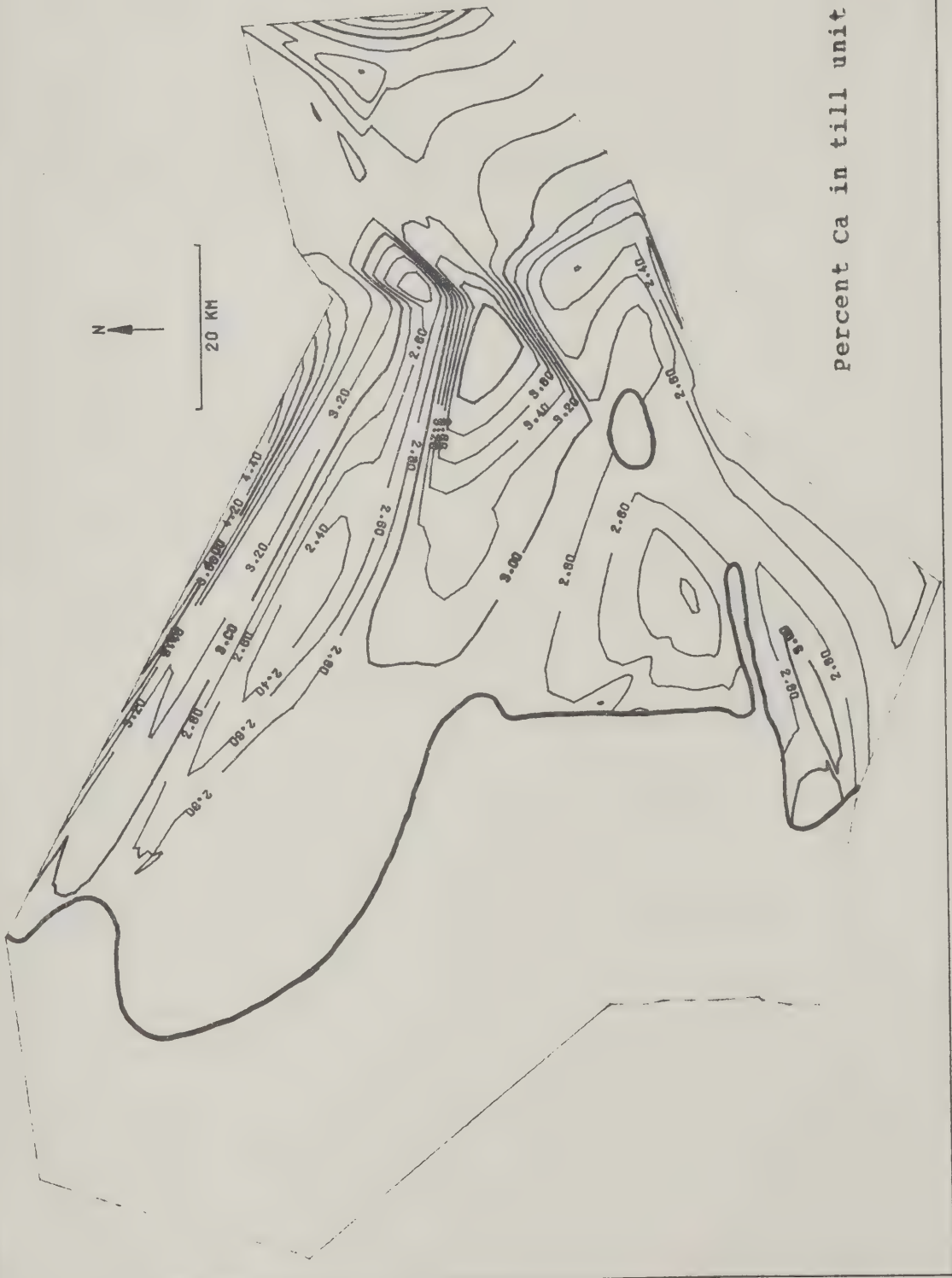




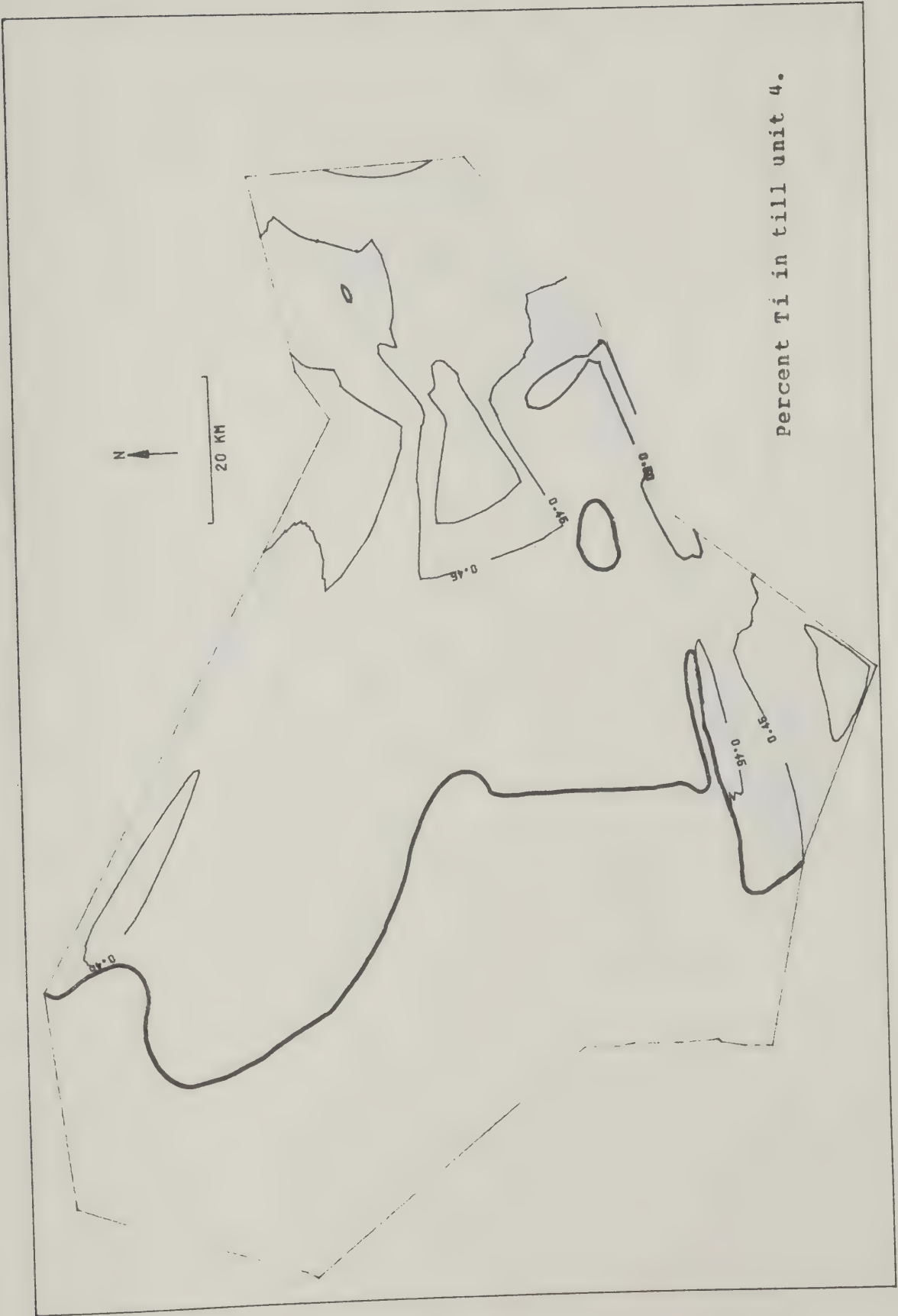




percent Ca in till unit 4.







Percent Ti in till unit 4.



Percent Fe in till unit 4.







## APPENDIX 8

Between-units versus within-units analysis of variance for each analyzed variable, with statistics describing the variation within each unit.



VARIABLE S1020 ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	6.4108	3.2054	4.635	0.0160
WITHIN UNITS	37	25.5871	0.6915		
TOTAL	39	31.9979			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	3.3752	0.7175	0.1647	1.7290	4.6176	3.0292 TO 3.7212
Unit 3	13	2.6166	0.8125	0.2253	1.7614	4.1787	2.1256 TO 3.1076
Unit 4	8	2.5124	1.0948	0.3871	1.2121	4.3533	1.5971 TO 3.4276
Total	40	2.9561	0.9058	0.1432	1.2121	4.6176	2.6664 TO 3.2458

VARIABLE S2040 ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	3.9570	1.9785	1.820	0.1762
WITHIN UNITS	37	40.2193	1.0870		
TOTAL	39	44.1763			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	4.5696	0.9349	0.2145	3.1437	6.6862	4.1190 TO 5.0201
Unit 3	13	4.6724	1.2309	0.3414	3.4471	7.0878	3.9286 TO 5.4162
Unit 4	8	3.8332	0.9492	0.3356	1.6953	4.5693	3.0396 TO 4.6267
Total	40	4.4557	1.0643	0.1683	1.6953	7.0878	4.1153 TO 4.7961



VARIABLE S4060 ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	5.926	2.9963	1.126	0.3351
WITHIN UNITS	37	98.4257	2.6602		
TOTAL	39	104.4183			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	7.2818	1.2574	0.2885	4.5444	10.4409	6.6758 TO 7.8879
Unit 3	13	7.3571	2.2832	0.6332	5.3412	11.7853	5.9774 TO 8.7368
Unit 4	8	8.2765	1.0288	0.3637	7.0418	9.6826	7.4164 TO 9.1366
Total	40	7.5052	1.6363	0.2587	4.5444	11.7853	6.9819 TO 8.0285

VARIABLE S60140 ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	3.9410	1.9705	0.313	0.7329
WITHIN UNITS	37	232.6135	6.2869		
TOTAL	39	236.5545			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	11.8355	2.8668	0.6577	8.0010	22.2258	10.4538 TO 13.2172
Unit 3	13	11.2991	2.5593	0.7098	8.3496	16.3881	9.7526 TO 12.8457
Unit 4	8	11.0991	0.9322	0.3296	10.0230	12.4416	10.3197 TO 11.8784
Total	40	11.5139	2.4628	0.3894	8.0010	22.2258	10.7262 TO 12.3015



VARIABLE S140230 ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	0.4643	0.2322	0.359	0.7010
WITHIN UNITS	37	23.9442	0.6471		
TOTAL	39	24.4085			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	5.1327	0.8237	0.1890	3.1628	6.2025	4.7357 TO 5.5297
Unit 3	13	5.3592	0.9325	0.2586	4.3495	7.3496	4.7957 TO 5.9228
Unit 4	8	5.1216	0.4305	0.1522	4.2522	5.7839	4.7617 TO 5.4816
Total	40	5.2041	0.7911	0.1251	3.1628	7.3496	4.9511 TO 5.4571

VARIABLE SAND ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	12.3845	6.1923	0.186	0.8312
WITHIN UNITS	37	1233.4270	33.3359		
TOTAL	39	1245.8115			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	32.1948	5.3981	1.2384	23.7607	50.1730	29.5930 TO 34.7965
Unit 3	13	31.3044	7.5355	2.0900	24.4220	46.7896	26.7507 TO 35.8581
Unit 4	8	30.8428	1.9824	0.7009	28.4426	33.6936	29.1855 TO 32.5001
Total	40	31.6350	5.6519	0.8936	23.7607	50.1730	29.8274 TO 33.4426





VARIABLE SILT ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	3.5209	1.7605	0.056	0.9454
WITHIN UNITS	37	1157.5150	31.2842		
TOTAL	39	1161.0359			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	56.0460	5.1831	1.1891	39.5737	65.1534	53.5478 TO 58.5441
Unit 3	13	56.2724	6.4721	1.7950	44.7264	66.8113	52.3613 TO 60.1834
Unit 4	8	55.4397	4.9468	1.7490	49.5110	61.4109	51.3041 TO 59.5754
Total	40	55.9983	5.4562	0.8627	39.5737	66.8113	54.2533 TO 57.7433

VARIABLE CLAY ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	21.6514	10.8257	1.380	0.2643
WITHIN UNITS	37	290.2926	7.8457		
TOTAL	39	311.9438			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	11.7591	1.7639	0.4047	8.9719	14.9542	10.9089 TO 12.6092
Unit 3	13	12.4230	3.4849	0.9666	7.0128	19.3107	10.3171 TO 14.5289
Unit 4	8	13.7174	3.5567	1.2575	9.5611	18.4745	10.7439 TO 16.6909
Total	40	12.3665	2.8282	0.4472	7.0128	19.3107	11.4620 TO 13.2710



VARIABLE LOCAL ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	85.0924	42.5462	4.006	0.0266
WITHIN UNITS	37	393.0085	10.6219		
TOTAL	39	478.1008			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	2.7999	2.1328	0.4893	0.0	7.0000	1.7719 TO 3.8279
Unit 3	13	5.7232	3.2010	0.8878	2.0000	13.5000	3.7889 TO 7.6576
Unit 4	8	5.7163	5.1847	1.8331	1.0638	16.5000	1.3818 TO 10.0508
TOTAL	40	4.3333	3.5013	0.5536	0.0	16.5000	3.2135 TO 5.4530

VARIABLE ACID ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	732.1665	366.0830	7.930	0.0014
WITHIN UNITS	37	1708.1632	46.1666		
TOTAL	39	2440.3296			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	68.2778	7.4608	1.7116	44.5000	77.0000	64.6818 TO 71.8738
Unit 3	13	59.8943	6.2235	1.7261	51.5000	68.6274	56.1335 TO 63.6552
Unit 4	8	59.4281	5.8728	2.0763	49.0000	67.5000	54.5183 TO 64.3378
Total	40	63.7833	7.9103	1.2507	44.5000	77.0000	61.2534 TO 66.3131



VARIABLE BASIC ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	45.6126	22.8063	0.557	0.5777
WITHIN UNITS	37	1515.3630	40.9557		
TOTAL	39	1560.9756			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	6.8051	8.6666	1.9883	2.0000	41.0000	2.6279 TO 10.9823
Unit 3	13	4.6493	2.3433	0.6499	1.5000	8.5000	3.2333 TO 6.0653
Unit 4	8	4.6956	3.7317	1.3194	1.0000	11.0000	1.5758 TO 7.8154
Total	40	5.6826	6.3265	1.0003	1.0000	41.0000	3.6592 TO 7.7059

VARIABLE CARB ANALYSIS OF VARIANCE

SOURCE	D.P.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	40.5804	20.2902	5.127	0.0108
WITHIN UNITS	37	146.4166	3.9572		
TOTAL	39	186.9970			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	4.4393	2.2312	0.5119	2.0000	8.5000	3.3639 TO 5.5147
Unit 3	13	4.1063	1.9289	0.5350	0.5000	7.5000	2.9407 TO 5.2719
Unit 4	8	1.8126	1.3182	0.4660	0.0	3.8095	0.7106 TO 2.9146
Total	40	3.8057	2.1897	0.3462	0.0	8.5000	3.1054 TO 4.5060



VARIABLE QTZ ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	841.9877	420.9937	14.984	0.0000
WITHIN UNITS	37	1039.5718	28.0965		
TOTAL	39	1881.5593			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	17.6778	4.1487	0.9518	10.5000	27.5000	15.6782 TO 19.6774
Unit 3	13	25.6267	5.3441	1.4822	16.5000	34.0000	22.3973 TO 28.8560
Unit 4	8	28.3473	7.4359	2.6290	13.5000	37.5000	22.1308 TO 34.5638
Total	40	22.3951	6.9459	1.0982	10.5000	37.5000	20.1737 TO 24.6165

VARIABLE CACO3 ANALYSIS OF VARIANCE

SOURCE	D.P.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	599.3153	299.6575	221.462	0.0000
WITHIN UNITS	37	50.0643	1.3531		
TOTAL	39	649.3794			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	12.6526	1.4975	0.3436	10.7000	16.6000	11.9308 TO 13.3744
Unit 3	13	5.7846	0.6902	0.1914	4.9000	7.0000	5.3675 TO 6.2017
Unit 4	8	3.8000	0.7540	0.2666	2.8000	5.1000	3.1696 TO 4.4304
Total	40	8.6500	4.0805	0.6452	2.8000	16.6000	7.3450 TO 9.9550





VARIABLE NA ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	0.4044	0.2022	7.782	0.0015
WITHIN UNITS	37	0.9615	0.0260		
TOTAL	39	1.3659			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	0.2677	0.1966	0.0451	0.0	0.5940	0.1730 TO 0.3625
Unit 3	13	0.1048	0.1386	0.0384	0.0	0.3620	0.0210 TO 0.1885
Unit 4	8	0.0250	0.0707	0.0250	0.0	0.2000	-0.0341 TO 0.0841
Total	40	0.1662	0.1871	0.0246	0.0	0.5940	0.1064 TO 0.2261

VARIABLE HG ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	10.4643	5.2321	239.223	0.0000
WITHIN UNITS	37	0.8092	0.0219		
TOTAL	39	11.2735			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	1.8580	0.1747	0.0401	1.6090	2.2940	1.7739 TO 1.9422
Unit 3	13	0.9070	0.1124	0.0312	0.7120	1.1000	0.8390 TO 0.9750
Unit 4	8	0.7340	0.1245	0.0440	0.6030	0.9790	0.6299 TO 0.8381
Total	40	1.3241	0.5376	0.0850	0.6030	2.2940	1.1522 TO 1.4961



VARIABLE AL ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	3.7789	1.8894	9.108	0.0006
WITHIN UNITS	37	7.6754	0.2074		
TOTAL	39	11.4543			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	7.6353	0.4702	0.1079	5.9090	8.0410	7.4087 TO 7.8619
Unit 3	13	8.2365	0.5383	0.1493	7.0250	8.7740	7.9112 TO 8.5618
Unit 4	8	8.2726	0.1769	0.0625	7.9270	8.4780	8.1247 TO 8.4205
Total	40	7.9582	0.5419	0.0857	5.9090	8.7740	7.7848 TO 8.1315

VARIABLE SI ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	23.3738	11.6869	62.613	0.0000
WITHIN UNITS	37	6.9062	0.1867		
TOTAL	39	30.2800			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	31.4125	0.3413	0.0783	30.9020	32.5930	31.2480 TO 31.5770
Unit 3	13	32.7265	0.5452	0.1512	32.0680	34.2280	32.3970 TO 33.0559
Unit 4	8	33.1990	0.4212	0.1489	32.5930	33.6860	32.8468 TO 33.5511
Total	40	32.1969	0.8811	0.1393	30.9020	34.2280	31.9151 TO 32.4787



VARIABLE K ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	0.2528	0.1264	15.229	0.0000
WITHIN UNITS	37	0.3071	0.0083		
TOTAL	39	0.5599			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	2.1440	0.1063	0.0244	1.8590	2.2970	2.0927 TO 2.1952
Unit 3	13	1.9839	0.0724	0.0201	1.8610	2.1140	1.9402 TO 2.0277
Unit 4	8	1.9862	0.0762	0.0269	1.9100	2.1490	1.9226 TO 2.0499
Total	40	2.0604	0.1198	0.0189	1.8590	2.2970	2.0221 TO 2.0987

VARIABLE CA ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	40.3753	20.1876	205.358	0.0000
WITHIN UNITS	37	3.6373	0.0983		
TOTAL	39	44.0125			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	4.6227	0.3474	0.0797	4.1690	5.4230	4.4552 TO 4.7901
Unit 3	13	2.8912	0.2476	0.0687	2.5600	3.3630	2.7416 TO 3.0408
Unit 4	8	2.2782	0.3227	0.1141	1.9300	2.9020	2.0084 TO 2.5481
Total	40	3.5911	1.0623	0.1680	1.9300	5.4230	3.2513 TO 3.9308



VARIABLE T1 ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	0.0843	0.0422	18.605	0.0000
WITHIN UNITS	37	0.0838	0.0023		
TOTAL	39	0.1682			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	0.4038	0.0433	0.0099	0.3350	0.4810	0.3830 TO 0.4247
Unit 3	13	0.4730	0.0419	0.0116	0.3950	0.5400	0.4477 TO 0.4983
Unit 4	8	0.5180	0.0644	0.0228	0.4420	0.6210	0.4641 TO 0.5719
Total	40	0.4491	0.0657	0.0104	0.3350	0.6210	0.4281 TO 0.4702

VARIABLE P2 ANALYSIS OF VARIANCE

SOURCE	D.F.	SUM OF SQUARES	MEAN SQUARES	F RATIO	F PROB.
BETWEEN UNITS	2	0.7204	0.3602	7.268	0.0022
WITHIN UNITS	37	1.8338	0.0496		
TOTAL	39	2.5542			

GROUP	COUNT	MEAN	STANDARD DEVIATION	STANDARD ERROR	MINIMUM	MAXIMUM	95 PCT CONF INT FOR MEAN
Unit 2	19	3.7473	0.2130	0.0489	2.9830	4.0150	3.6446 TO 3.8499
Unit 3	13	3.9827	0.2408	0.0668	3.5040	4.3810	3.8372 TO 4.1282
Unit 4	8	4.0567	0.2145	0.0758	3.7760	4.3530	3.8775 TO 4.2360
Total	40	3.8857	0.2559	0.0405	2.9830	4.3810	3.8038 TO 3.9675





## APPENDIX 9

Analysis of variance for each analyzed variable, comparing variation between areas, till units and vertical positions within till units.

The western test area contains drill holes 435E, 436E, 437E and 758E, the central area contains holes 456E, 498E and 709E, while the eastern area contains holes 457E, 458E and 765E.



SOURCE OF S1020 VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	6.593	2.	3.295	2.614	0.142
H within A	8.824	7.	1.261		
Within Drill Holes					
Units (U)	3.134	1.	3.134	2.709	0.144
U x A	20.764	2.	10.382	8.974	0.012
U x H within A	8.098	7.	1.157		
Position (P)					
P x A	2.772	1.	2.772	3.312	0.112
P x H within A	2.053	2.	1.026	1.226	0.349
	5.859	7.	0.837		
U x P	0.113	1.	0.113	0.273	0.617
U x P x A	1.810	2.	0.905	2.188	0.183
U x P x H within A	2.896	7.	0.414		

SOURCE OF S2040 VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	4.560	2.	2.280	0.670	0.542
H within A	23.836	7.	3.405		
Within Drill Holes					
Units (U)	9.605	1.	9.605	3.430	0.106
U x A	37.664	2.	18.832	6.725	0.023
U x H within A	19.602	7.	2.800		
Position (P)					
P x A	10.491	1.	10.491	3.916	0.088
P x H within A	5.114	2.	2.557	0.955	0.430
	18.753	7.	2.679		
U x P	1.346	1.	1.346	0.839	0.390
U x P x A	2.541	2.	1.270	0.792	0.430
U x P x H within A	11.229	7.	1.604		



SOURCE OF S4060 VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	8.672	2.	4.336	1.567	0.274
H within A	19.370	7.	2.767		
Within Drill Units (U)	13.206	1.	13.206	13.003	0.009
U x A	40.967	2.	20.484	20.169	0.001
U x H within A	7.109	7.	1.016		
Position (P)	0.049	1.	0.049	0.021	0.888
P x A	2.437	2.	1.218	0.529	0.611
P x H within A	16.136	7.	2.305		
U x P	22.601	1.	22.601	6.730	0.036
U x P x A	21.727	2.	10.863	3.235	0.101
U x P x H within A	23.507	7.	3.358		

SOURCE OF S60140 VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	34.791	2.	17.396	0.474	0.641
H within A	257.066	7.	36.724		
Within Drill Units (U)	12.472	1.	12.472	0.936	0.366
U x A	124.371	2.	62.185	4.667	0.052
U x H within A	93.273	7.	13.325		
Position (P)	4.213	1.	4.213	0.222	0.652
P x A	41.318	2.	20.659	1.089	0.387
P x H within A	132.754	7.	18.965		
U x P	1.674	1.	1.674	0.170	0.692
U x P x A	60.649	2.	30.324	3.087	0.109
U x P x H within A	68.758	7.	9.823		



SOURCE OF S140230 VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	40.987	2.	20.493	0.963	0.426
H within A	148.260	7.	21.180		
Within Drill Holes					
Units (U)	19.014	1.	19.014	1.021	0.346
U x A	0.440	2.	0.220	0.012	0.988
U x H within A	130.391	7.	18.627		
Position (P)	4.067	1.	4.067	0.369	0.563
P x A	25.185	2.	12.593	1.142	0.372
P x H within A	77.214	7.	11.031		
U x P	0.003	1.	0.003	0.000	0.988
U x P x A	22.856	2.	11.428	0.851	0.467
U x P x H within A	94.052	7.	13.436		

SOURCE OF SAND VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	68.165	2.	34.082	1.275	0.337
H within A	187.141	7.	26.734		
Within Drill Holes					
Units (U)	87.085	1.	87.085	2.858	0.135
U x A	796.104	2.	398.052	13.061	0.004
U x H within A	213.328	7.	30.475		
Position (P)	21.592	1.	21.592	1.664	0.238
P x A	43.607	2.	21.803	1.680	0.254
P x H within A	90.840	7.	12.977		
U x P	25.082	1.	25.082	1.090	0.331
U x P x A	216.920	2.	108.460	4.715	0.050
U x P x H within A	161.031	7.	23.004		





SOURCE OF SILT VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	62.259	2.	31.129	0.730	0.515
H within A	298.688	7.	42.670		
Within Drill Holes					
Units (U)	9.320	1.	9.320	2.416	0.164
U x A	309.592	2.	154.796	40.132	0.001
U x H within A	27.000	7.	3.857		
Position (P)					
P x A	38.697	1.	38.697	4.691	0.067
P x H within A	33.776	2.	16.888	2.047	0.200
	57.750	7.	8.250		
U x P	7.952	1.	7.952	0.293	0.605
U x P x A	31.551	2.	15.776	0.581	0.584
U x P x H within A	190.000	7.	27.143		

SOURCE OF CLAY VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	37.410	2.	18.705	0.698	0.529
H within A	187.531	7.	26.790		
Within Drill Holes					
Units (U)	153.364	1.	153.364	7.695	0.028
U x A	112.903	2.	56.452	2.833	0.126
U x H within A	139.508	7.	19.930		
Position (P)					
P x A	2.476	1.	2.476	0.389	0.553
P x H within A	36.126	2.	18.063	2.836	0.125
	44.578	7.	6.368		
U x P	4.784	1.	4.784	0.633	0.452
U x P x A	88.960	2.	44.480	5.889	0.032
U x P x H within A	52.875	7.	7.554		



SOURCE OF LOCAL VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	55.684	2.	27.842	1.909	0.218
H within A	102.102	7.	14.586		
Within Drill Holes					
Units (U)	2.294	1.	2.294	0.121	0.738
U x A	85.319	2.	42.660	2.257	0.175
U x H within A	132.303	7.	18.900		
Position (P)	2.157	1.	2.157	0.116	0.743
P x A	7.871	2.	3.935	0.212	0.814
P x H within A	130.021	7.	18.574		
U x P	1.273	1.	1.273	0.044	0.839
U x P x A	52.329	2.	26.165	0.912	0.445
U x P x H within A	200.749	7.	28.678		

SOURCE OF ACID VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	1571.101	2.	785.550	5.474	0.037
H within A	1004.625	7.	143.518		
Within Drill Holes					
Units (U)	3.899	1.	3.899	0.029	0.870
U x A	503.744	2.	251.872	1.874	0.223
U x H within A	941.000	7.	134.429		
Position (P)	65.045	1.	65.045	1.977	0.202
P x A	42.239	2.	21.119	0.642	0.555
P x H within A	230.250	7.	32.893		
U x P	66.822	1.	66.822	0.387	0.554
U x P x A	125.974	2.	62.987	0.364	0.707
U x P x H within A	1210.063	7.	172.866		



SOURCE OF BASIC VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	45.955	2.	22.977	0.686	0.535
H within A	234.521	7.	33.503		
Within Drill Holes					
Units (U)	3.675	1.	3.675	0.100	0.761
U x A	49.855	2.	24.727	0.672	0.541
U x H within A	257.764	7.	36.823		
Position (P)	17.139	1.	17.139	0.642	0.449
P x A	71.770	2.	35.885	1.345	0.320
P x H within A	186.738	7.	26.677		
U x P	3.001	1.	3.001	0.065	0.806
U x P x A	49.023	2.	24.512	0.533	0.609
U x P x H within A	322.191	7.	46.027		

SOURCE OF CARB VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	1.688	2.	0.844	0.010	0.990
H within A	595.837	7.	85.120		
Within Drill Holes					
Units (U)	3.588	1.	3.588	0.035	0.857
U x A	280.569	2.	140.285	1.358	0.317
U x H within A	723.213	7.	103.316		
Position (P)	22.500	1.	22.500	0.673	0.439
P x A	41.245	2.	20.622	0.617	0.566
P x H within A	233.853	7.	33.408		
U x P	206.041	1.	206.041	6.979	0.033
U x P x A	30.637	2.	15.319	0.519	0.616
U x P x H within A	206.651	7.	29.522		



SOURCE OF Q <sub>12</sub> VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	P PROBABILITY
Between Drill Holes (H)					
Area (A)	770.893	2.	385.447	1.435	0.300
H within A	1879.742	7.	268.534		
Within Drill Holes					
Units (U)	18.243	1.	18.243	0.284	0.611
U x A	242.386	2.	121.193	1.885	0.221
U x H within A	450.008	7.	64.287		
Position (P)	35.911	1.	35.911	0.314	0.593
P x A	1.406	2.	0.703	0.006	0.994
P x H within A	799.754	7.	114.251		
U x P	46.023	1.	46.023	0.457	0.521
U x P x A	22.781	2.	11.391	0.113	0.895
U x P x H within A	704.543	7.	100.649		

SOURCE OF CAGOS VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	P PROBABILITY
Between Drill Holes (H)					
Area (A)	85.334	2.	42.667	22.532	0.001
H within A	13.255	7.	1.894		
Within Drill Holes					
Units (U)	10.882	1.	10.882	2.094	0.191
U x A	13.203	2.	6.601	1.270	0.338
U x H within A	36.379	7.	5.197		
Position (P)	0.055	1.	0.055	0.047	0.835
P x A	1.282	2.	0.641	0.541	0.605
P x H within A	8.296	7.	1.185		
U x P	1.222	1.	1.222	3.241	0.115
U x P x A	1.923	2.	0.962	2.551	0.147
U x P x H within A	2.639	7.	0.377		





SOURCE OF NA VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	0.330	2.	0.165	4.375	0.059
H within A	0.264	7.	0.038		
Within Drill Holes					
Units (U)	0.002	1.	0.002	0.015	0.906
U x A	0.093	2.	0.046	0.370	0.704
U x H within A	0.876	7.	0.125		
Position (P)	0.252	1.	0.252	11.903	0.011
P x A	0.087	2.	0.043	2.049	0.199
P x H within A	0.148	7.	0.021		
U x P	0.013	1.	0.013	0.551	0.482
U x P x A	0.018	2.	0.009	0.381	0.696
U x P x H within A	0.161	7.	0.023		

SOURCE OF HG VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	0.763	2.	0.381	2.938	0.118
H within A	0.908	7.	0.130		
Within Drill Holes					
Units (U)	0.308	1.	0.308	3.471	0.105
U x A	0.047	2.	0.023	0.264	0.775
U x H within A	0.621	7.	0.089		
Position (P)	0.190	1.	0.190	3.608	0.099
P x A	0.022	2.	0.011	0.206	0.819
P x H within A	0.369	7.	0.053		
U x P	0.063	1.	0.063	2.682	0.145
U x P x A	0.142	2.	0.071	3.028	0.113
U x P x H within A	0.165	7.	0.024		



SOURCE OF AL VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	0.261	2.	0.131	0.409	0.680
H within A	2.239	7.	0.320		
Within Drill Holes					
Units (U)	0.002	1.	0.002	0.010	0.922
U x A	0.618	2.	0.309	1.341	0.321
U x H within A	1.614	7.	0.231		
Position (P)	0.009	1.	0.009	0.122	0.737
P x A	0.222	2.	0.111	1.540	0.279
P x H within A	0.505	7.	0.072		
U x P	0.004	1.	0.004	0.035	0.856
U x P x A	0.682	2.	0.341	3.023	0.113
U x P x H within A	0.790	7.	0.113		

SOURCE OF SI VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	4.065	2.	2.033	2.448	0.156
H within A	5.813	7.	0.830		
Within Drill Holes					
Units (U)	0.230	1.	0.230	0.235	0.643
U x A	0.665	2.	0.332	0.339	0.723
U x H within A	6.855	7.	0.979		
Position (P)	1.112	1.	1.112	2.094	0.191
P x A	0.115	2.	0.058	0.108	0.899
P x H within A	3.719	7.	0.531		
U x P	0.217	1.	0.217	0.371	0.562
U x P x A	0.205	2.	0.102	0.174	0.844
U x P x H within A	4.105	7.	0.586		



SOURCE OF K VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	0.161	2.	0.080	2.066	0.197
H within A	0.272	7.	0.039		
Within Drill Holes					
Units (U)	0.014	1.	0.014	0.592	0.433
U x A	0.005	2.	0.002	0.120	0.888
U x H within A	0.141	7.	0.020		
Position (P)	0.011	1.	0.011	0.359	0.568
P x A	0.010	2.	0.005	0.167	0.849
P x H within A	0.211	7.	0.030		
U x P	0.000	1.	0.000	0.021	0.888
U x P x A	0.001	2.	0.000	0.020	0.980
U x P x H within A	0.115	7.	0.016		

SOURCE OF CA VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	PROBABILITY
Between Drill Holes (H)					
Area (A)	5.185	2.	2.593	7.236	0.020
H within A	2.508	7.	0.358		
Within Drill Holes					
Units (U)	0.180	1.	0.180	0.340	0.578
U x A	0.515	2.	0.258	0.487	0.634
U x H within A	3.700	7.	0.529		
Position (P)	0.522	1.	0.522	1.806	0.221
P x A	0.416	2.	0.208	0.721	0.519
P x H within A	2.022	7.	0.289		
U x P	0.119	1.	0.119	0.740	0.418
U x P x A	0.490	2.	0.245	1.529	0.281
U x P x H within A	1.122	7.	0.160		



SOURCE OF TI VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	P PROBABILITY
Between Drill Holes (H)					
Area (A)	0.012	2.	0.006	1.715	0.248
H within A	0.025	7.	0.004		
Within Drill Holes					
Units (U)	0.002	1.	0.002	1.137	0.322
U x A	0.009	2.	0.004	2.091	0.194
U x H within A	0.014	7.	0.002		
Position (P)	0.003	1.	0.003	3.063	0.124
P x A	0.001	2.	0.001	0.693	0.531
P x H within A	0.007	7.	0.001		
U x P	0.004	1.	0.004	0.825	0.394
U x P x A	0.020	2.	0.010	2.163	0.186
U x P x H within A	0.032	7.	0.005		

SOURCE OF FE VARIATION	SUM OF SQUARES	DEGREES OF FREEDOM	MEAN SQUARES	F RATIO	P PROBABILITY
Between Drill Holes (H)					
Area (A)	0.045	2.	0.023	0.162	0.854
H within A	0.983	7.	0.140		
Within Drill Holes					
Units (U)	0.121	1.	0.121	0.705	0.429
U x A	0.262	2.	0.131	0.761	0.502
U x H within A	1.205	7.	0.172		
Position (P)	0.159	1.	0.159	3.412	0.107
P x A	0.120	2.	0.060	1.287	0.334
P x H within A	0.327	7.	0.047		
U x P	0.014	1.	0.014	0.130	0.729
U x P x A	0.159	2.	0.080	0.737	0.512
U x P x H within A	0.756	7.	0.108		

















**B30207**